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Late-Hercynian Intrusion-related gold deposits: an integrated model on the Tighza polymetallic district, central Morocco

Revised version for Journal of African Earth Sciences

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Abstract

Gold have been recently recognized in the Tighza (formerly Jebel Aouam) district, in the Hercynian belt of central Morocco. This district has long been known for its W mineralization, as well as major Pb-Ag-Zn, and minor Sb-Ba deposits, all geographically associated with late-Hercynian calc-alkaline magmatism. Gold mineralization in the district is mainly hosted by thick W-Au quartz veins located around the “Mine granite” small granitic plug. Within the veins, gold grade is highest (up to 70 g/t) close to the granite but rapidly decreases going outward from the granite, defining a perigranitic zoning. Anomalous gold grades have also been measured in hydrothermal skarn layers close to two other granitic plugs (Kaolin granite and Mispickel granite), associated with disseminated As-Fe sulfides. The paragenetic sequence for the W-Au quartz veins shows three stages: 1) an early oxidized stage with wolframite-scheelite associated with early quartz (Q₁), 2) an intermediate Bi-As-Te-Mo-Au sulfide stage with loellingite, bismuth minerals and native gold with a later quartz (Q₂), restricted to a narrow distance from the granite, and 3) a late lower temperature As-Cu-Zn-(Pb) stage with abundant massive pyrrhotite, arsenopyrite and sphalerite, locally forming independent veins (“pyrrhotite vein”). Both Q₁ hyaline and Q₂ saccharoidal gold-bearing quartz display aqua-carbonic fluids with minor H₂S and Cu and an homogeneous composition (81 mole % H₂O, 18 mole % CO₂ and about 1 mole % NaCl). The trapping pressure is estimated to 1.5 to 2 kbar with temperature ranging from 300° to 350°C. Q₁ inclusions have exploded indicating an uplift of the Tighza block, that lead to saccharoidal Q₂ quartz deposition with multiphase NaCl-saturated fluid inclusions. ⁴⁰Ar/³⁹Ar dating demonstrates that the “Mine granite”, tungsten skarnoid, scheelite-molybdenite veins, and very likely gold-bearing veins are coeval, emplaced at 286 ± 1 Ma. Multiple and widespread metal sources are indicated by radiogenic isotope studies. Nd and Sr isotope compositions of scheelite and granites

suggest the participation of a juvenile component while lead isotopes demonstrate a major participation of the basement.

Both gold mineralization and zoning suggest that the system developed at the end of the magmatic activity, accompanying a major transition in magmatic fluid composition. The morphology of the gold-bearing mineralization is dependent of the permeability and the reactivity of host-rocks: focus circulation of fluids through pre-existing tectonic corridors, reactivated by late-Hercynian intrusions favor the formation of large W-type gold veins, while infiltration of fluid within reactive stratigraphic layers gives rise to skarn mineralization. A $^{40}\text{Ar}/^{39}\text{Ar}$ date (W1 north vein: 291.8 ± 0.3 Ma) indicates that hydrothermal circulation predates gold and tungsten deposition in open fractures as well as Mine granite emplacement.

The W-Au mineralization preceded the onset of a large convective hydrothermal cell around the intrusion that led to the formation of the Pb-Ag-Zn mined veins. The Tighza polymetallic district displays numerous similarities with the R-IRG model that was defined in the American Cordillera, such as thermal and zonation patterns, carbonic hydrothermal fluids and chronology of intrusion and related deposits, but also provides new insight to the R-IRG model such as wide Au-quartz veins instead of sheeted Au-veins, oxidation state of the magma, and Sr-Nd isotopic data. These results establish a major magmatic contribution and discard a direct genetic relationship between gold mineralization and major neighboring Pb-Ag-Zn veins. A large number of classic Pb-Zn district of the Western Hercynides belong to the same clan.

Keywords: gold, R-IRG, Tighza, Morocco, Hercynian, perigranitic

1 - Introduction

The strong demand for gold and the maturity of several major mining districts have prompted new research on gold deposits associated with plutonic systems. Three main gold deposit types include: (1) porphyry copper-gold (Sillitoe, 2000); (2) Reduced intrusion-related gold (Lang and Baker, 2001); and (3) Iron Oxide Copper Gold deposits (Groves et al., 2010). Reduced Intrusion-related gold deposits (R-IRG) have been mainly described in the American Cordillera, and especially in the Tintina belt, such as the Fort Knox and Dublin Gulch deposits (Hollister, 1992; Sillitoe and Thompson, 1998; Thompson et al., 1999, Hart, 2005). They were previously described under the name of Thermal Aureole Gold (Wall and Taylor, 1991). The R-IRG type has been extended to the late Paleozoic Tianshan deposits, such as Muruntau, Vasilkovskoye, Amantaitau, and Kumtor), not without some controversies (Mao et al., 2004; Morelli et al., 2007). Several European gold deposits of the Hercynian belt have also been associated with the R-IRG type, such as the Mokskro (Czech Republic), Salsigne (France) and Rio Narcea (Spain) (Lang et al., 2000), and define a major metallogenic event around 300 Ma (Gloaguen et al., 2003; Bouchot et al., 2005).

R-IRG deposits are characterized by: (1) a location within a deformed shelf sequence; (2) an association with relatively reduced plutons, of calc-alkaline composition, transitional between ilmenite and magnetite series with Sn-W mineralization; (3) a post-deformation emplacement, like orogenic gold deposits; (4) a low sulfide content, and 5) an association of proximal gold ore with distal base metal-rich veins. Fluids associated with the gold mineralization are CO₂-rich hydrothermal fluids carrying Au, Bi, W, As, Mo, Te, and/or Sb (Hart, 2005). The genesis of these deposits remains, however, poorly understood. The limit of this new family of deposits, as the concept is more and more used outside its early definition. Major uncertainties remain concerning the genetic link between intrusion and mineralization including: i) the origin of fluids and

elements, coming from the mantle or the crust, derived from magmas or by convection in the host rocks, ii) the thermal role of intrusion and the duration of the hydrothermal stage; iii) the connection between gold and associated ore deposits. The main objective of this paper is to better constrain the R-IRG model based on the example of the classic lead-zinc-silver Tighza mining district (formerly Jebel Aouam, Central Morocco Massif) where gold was recently discovered.

The Central Morocco Massif is the largest outcropping segment of the Hercynian belt in Morocco. It is part of the large Variscan belt of the Circum-Atlantic regions extended over 6 000 km long between the Laurasia and Gondwana super-continent (Michard et al., 2010). It displays mainly Paleozoic rocks folded and locally thrust toward the West, cut by Late Carboniferous granitic massifs. Numerous mineral deposits have been mined, including tin and tungsten (Oulmes; Tahiri et al., 2007), fluorite (El Hammam), and lead-zinc-silver in Tighza (Agard et al., 1958; Jébrak, 1984; Cheilletz, 1984). After more than 40 years of Pb-Ag-Zn mining (6 Mt @ 8 % Pb, plus Ag and Zn; Wadjinny, 1998), gold was discovered in the Tighza district using drillings and limited underground workings. The Tighza gold mineralization provides details of the early metallogenic processes that occur in association with calc-alkaline plutonism in an I-IRG system. These metallogenic processes are studied by integrating field observations, detailed mineralogy, fluid inclusion and isotopic data within the structural evolution framework of the area at the end of the pluton emplacement.

2 - Geology and ore deposits of the Tighza district

2.1 - Geology

The Central Morocco Massif belongs to the Meseta domain of the Hercynian belt. This domain results from the Late Devonian–Late Carboniferous accretion of several Paleozoic sub-domains

that reflect the development of a passive margin on the edge of the African craton during Early Paleozoic. Tighza is located in Ordovician to Carboniferous sedimentary units of the Nappe zone of eastern Central Morocco (Piqué et al., 1993). Tectonic shortening of the units during the Late Visean–Early Namurian occurred in association with the emplacement of gravity-driven synsedimentary nappes. Tectonic vergency is dominantly toward the WNW, with upright or NW-verging folds associated with flat-ramp faults (Michard et al., 2010).

The Central Morocco Massif is crosscut by dextral transpressional NE-SW faults of Carboniferous to Permian in age that controlled the location of granodioritic intrusions and subsequently, the location of detrital basins. The granodioritic intrusions are interpreted to have formed from mixing between anatectic magmas derived from partial melting of continental crust and mantle-derived magmas (Gasquet et al., 1996; El Hadi et al., 2006). The intrusions were emplaced at supracrustal levels as late-tectonic elongated massifs between the Visean–Autunian period (ca 330 to 300–290 Ma). Late activation on the NE-SW faults resulted in the formation of pull-apart basins (Saidi et al., 2002) and associated alkaline felsic volcanism (Mrini et al., 1992). To our knowledge, no low-angle normal fault accommodating the collapse of the orogeny has yet been recognized.

In the Tighza area, upper Ordovician units are mainly detrital and composed of micaceous shales, with beds of sandstone, grading into quartzite units associated with fine conglomerates. Silurian sequences are composed of clay-rich detrital units and limestone. The amount of calcareous sedimentation increased during Early and Middle Devonian time. Above a regional unconformity (Fig. 1), Tournaisian sandstones (360 - 350 Ma) conglomerates, and greywackes are overlain by marine sedimentary rocks including Middle Visean bioclastic limestones and marls (~ 340 Ma) and a thick pile of Visean-age flysch.

Figure 1- Location and geology of the Tighza mining district with its various ore deposits

Late to post-Hercynian felsic intrusions can be divided in three groups (Cheilletz, 1984, Ntarmouchant, 1991): (1) an early calc-alkaline magmatic suite, composed of dikes of microgranodiorite, microgranite and rhyolite, with biotite granitic/granodioritic stocks, dated between 293.0 ± 3.0 and 287.9 ± 5.1 Ma; (2) late tholeiitic microdioritic and microtonalitic dikes; and (3) sparse leucogranitic dikes.

Granitic stocks of the first group are mainly represented by three small bodies, about 1 km² each, called from north to south: Mispickel granite (also named Bou-Iqualouchène), Mine granite, and Kaolin granite (Fig. 1). Although they appear slightly different in composition, these granitic stocks are considered as apexes of a larger magmatic body at depth (Agard et al., 1958; Eldursi, 2009; Branquet et al., in prep.). A broad zone of contact metamorphism (3 x 2 km on surface, Fig. 1) includes the three granite stocks supporting the presence of a unique magmatic body at depth.

Contact metamorphism is marked by the transformation of pelitic sediments to spotted hornfels showing a quartz, biotite, muscovite, andalusite and cordierite association. These mineral assemblages suggest a maximum temperature of 650°C, with a lithostatic pressure around 2 kbar (Cheilletz and Giuliani, 1988). Locally, small-sized skarns are enriched in scheelite in Visean and Ordovician carbonate layers.

The low $\text{Fe}^{3+}/(\text{Fe}^{3+}+\text{Fe}^{2+})$ ratios for the three granite (Table 1), their relative high modal amount of primary sulfides and the lack of primary magnetite (Nerci, 2006), suggest that they belong to the S-type granites of the ilmenite series (Whalen and Chapell, 1988). However, on a regional scale, Giuliani et al. (1989) show that the Jebel Aouam plutons, such as the Zaer biotite granodiorite unit

west of the Tighza area, have geochemical characteristics characteristic of a mixture of felsic (crustal) and mafic (mantle) sources. Based on geochemistry and other arguments, Giuliani et al. (1989) qualified this calc-alkaline suite as a “post collisional, uplift environment” type.

Tholeiitic microdioritic dikes of the second group crosscut the W mineralized veins without movement along strike, but do not cross the Pb-Ag-Zn veins. Their orientation is compatible with the late tectonic event recorded throughout the area (D3 event, see below). The dikes correspond to a tholeiitic series with a calc-alkaline component that has no clear genetic relationships with other magmatic rocks of the Moroccan Central Massif (Ntarmouchant, 1991).

Dikes of leucogranites from silico-aluminous magmatic suites have been reported in the Tighza area (Ntarmouchant, 1991). These dikes postdate the calc-alkaline suite and are likely correlated to numerous and widespread two-mica granites in Central Morocco (e.g. the Zaer two-mica granite unit, the peraluminous Oulmès granite; Giuliani et al., 1989) derived from partial melting of metasediments after crustal thickening (i.e. S-type granites).

The structural evolution of the district has been deciphered by Desteucq (1974), Cheilletz (1983), and Branquet et al. (in prep.). Three main deformation events can be distinguished:

- i) A thin-skinned event (D1) leading to the stacking of an allochthonous thrust sheet unit to the east over a relative autochthonous unit to the west (Fig. 1). Sparse stretching lineations along the sole thrust suggest that the transport direction is NE-SW;
- ii) A ductile-brittle event associated with a NW-SE horizontal shortening (D2). This deformation is responsible for the NE-SW folding and related cleavage development which characterizes the present-day structure of the area. Reverse limbs also formed

during this shortening event. E-W fractures and veins (such as the W1) and the major Tighza fault (Fig. 1) formed as oblique-slip dextral normal faults and hybrid extensional-shear- fractures (Branquet et al. in prep.). Tighza granitic stocks are associated with the D2 event, and thus D2 happened at about 290 Ma.

- iii) Stress variations lead to a late brittle transtensive left-lateral wrenching event (D3) that is documented along the Tighza fault and compatible with the formation of the Pb-Ag-Zn vein system (Desteucq, 1974; Cheilietz, 1983). A $^{40}\text{Ar}/^{39}\text{Ar}$ age of 205 ± 1 Ma is indicated based on dating of contemporaneous major fluorite deposits in the region such as El Hammam (Cheilietz et al., 2010). However, in the Central Morocco Massif, isotopic systems have been strongly reset by several thermal events related to the opening of the Atlantic Ocean (Jébrak, 1984; Ghorbal et al., 2008; Barbero et al., 2011) invoking a note caution in the interpretation of radiometric ages of weak minerals associated with the vein mineralization.

2.2 - Mineral Deposits

The Tighza district displays three types of mineral deposits: i) W-Au occurrences in the core of the district; ii) Widely distributed and exploited Pb-Ag-Zn veins (Sidi Ahmed, Signal, Ighem Aousser...); and iii) Sb-Ba veins, located at the periphery of the district (Figs. 1 and 2).

Figure 2 – Gold and tungsten-bearing structures of the Tighza mining district. See text for discussion

Several large Pb-Ag-Zn vein deposits (Signal, Ighem Ousser, Sidi Ahmed...) striking N50°E to N80°E and dipping N to NW (average 65°) collectively contain more than 11 Mt of ore, with 1 Mt Pb and more than 1,000 t Ag (Wadjinny, 1998), giving a grade of 9% Pb and 90 g/t Ag ; zinc is now recovered since few years. They also yield significant amounts of zinc that was not recovered. The veins extend for more than 6 km, with an average thickness of 2 m, displaying pinch-and-swell structures related to transtensional faulting. These veins are hosted by Paleozoic sediments, and cut across the granitic stocks (Kaolin and Mine granites) as well as tungsten veins (Desteucq, 1974). They are associated with conjugate strike-slip faults resulting from the NE-SW shortening (i.e. D3 event). An $^{40}\text{Ar}/^{39}\text{Ar}$ age of 267.7 ± 2.3 Ma was measured by Watanabe (2002) on hydrothermal illite from the Ighem Aousser vein. Two successive mineralization stages have been recognized: an early stage with siderite, quartz, silver-bearing galena, sphalerite, chalcopryrite, pyrite and traces of Ni, Sn and Bi minerals, and a later stage with calcite, chalcedony, barite and less silver-rich galena, often exhibiting a banded texture, and collapse breccias (Jébrak, 1984).

Small silicified stibnite veins with barite are hosted by Paleozoic sediments in the outer parts of the district. They have been actively mined in the past, but are now not accessible. Dissemination of stibnite also appears in association within felsic dikes.

Tungsten-bearing mineral occurrences have been described by Cheilietz (1984) and Cheilietz and Isnard (1985) within the metamorphic aureole of the mine granite. Two types were distinguished (Figs. 1, 2 and 3):

- Stratabound, lens-shaped, skarnoid deposits in Ordovician terranes, with scheelite, Mg-biotite and quartz, only known at depth (Figs. 3 c and d): The association comprises actinolite, plagioclase, K-feldspar, pyrrhotite, chalcopryrite, ilmenite, sphalerite and titanite (Cheilietz and Giuliani, 1988). These skarns were interpreted to have resulted from

metasomatic process induced by magmatic fluids (Cheilletz, 1984); they are crosscut by small dykes of diorite-tonalite of the late magmatic group, reflecting their early emplacement;

- Vein-type deposits that cross-cut the skarnoid deposits (Figs. 2 and 3) such as W1 tungsten and gold vein (see below).

Biotite from the W1 vein yielded $^{40}\text{Ar}/^{39}\text{Ar}$ ages of 288.4 ± 3.1 Ma and 285 ± 3.8 Ma whereas biotite from the stratabound lens-shaped skarnoids gave ages of 285.3 ± 3 Ma and 284.1 ± 4.2 Ma (Cheilletz et Zimmermann, 1982; Cheilletz, 1984), demonstrating their association with the first magmatic stage.

The T-P conditions of emplacement of these tungsten mineralization events were determined by fluid inclusion study to be 580°C and 1.8 kbar for the stratabound lens-shaped skarnoids, and 400° to 450°C and 1 kbar for vein-style deposit (Cheilletz and Isnard, 1985). Skarnoids fluids were of moderate to high salinity (6-18 wt % NaCl equiv.; Cheilletz and Giuliani, 1988).

3 - Geology and mineralogy of gold deposits

Although tungsten-bearing veins have been documented since mineral exploration of the district began 60 years ago, the potential for gold has only been recognized over the past 20 years. Gold occurrences have been discovered in i) tungsten-bearing quartz veins (the “W veins”, ii) quartz-scheelite stockwork, and iii) As-Fe-(Au) dissemination in Visean skarn (Figs. 2, 3).

3.1 - Au-W quartz veins

Au-W-bearing veins are found immediately adjacent to the Mine granite. They form a regular, locally anastomosed, network, with a strike varying from N 90°E to N 120°E. Major veins include W1, W1 north, WO3, W4, W5 and W6 (Figs. 2, 3a). W1 north and W4 have been intersected in drill hole at depths of up to 220 m. Their thickness reaches locally 2 m. Two minor arsenopyrite veins, located at the southwestern edge of the Mispickel Granite, and a decimetric quartz-scheelite-bismuth-molybdenite vein at level 707 m belong probably to the same W vein system (Fig. 2).

The largest vein system is the W1-W1 north vein that can be followed over a distance of 1400 m. It dips 50°N to 85°N, and displays a large virgation toward the north in its western part. From the contact with the Mine granite, the fracture continues within the granite. It appears as a microgranite dike with rhyolitic texture that was subsequently strongly silicified and sericitized and now appears as a greisen (Fig. 3f). This zone is meters thick, strikes N130°E, and shows a significant gold enrichment (>0.4 ppm). The W1-W1 north vein system crosscuts rhyolitic dykes but is itself cut by dark tonalite dykes, and several dextral and sinistral wrench faults. On the south-east side of the Mine granite, two other veins (W4 and W5) display similar characteristics with a slightly different strike (N125°E for W4). The complex geometry of the W veins is indicative of a transtensional dextral fault systems with a horsetail pattern, a structure suggestive of fracture reactivation during NW-SE shortening.

The W veins are mainly composed of massive hyaline quartz, becoming white grey (Q₁) and then saccharoidal (Q₂) due to tectonic activity and new silica input. Quartz is associated with muscovite (W1-W1north) or biotite, phlogopite (WO3, W4 and W5), sulfides and native gold. Detailed gold analyses of the vein demonstrate a strong zoning: highest grades occur immediately near the Mine granite (up to 50 ppm), and decrease outward: 9 ppm from the contact to 200 m, 3 ppm at 400 m

(mean values) (Fig. 4). The highest grades have been measured in the W1 north near the granite (average grade of 4.36 ppm on 500 m in surface, and 9.3 ppm on 363 m starting from the granite), and in the WO3 (average 166 ppm).

Figure 3 – Rock Field exposures in Tighza mining district

a – W1 north Au-W quartz vein. Overburden was removed to allow a precise grab sampling, yielding the highest gold values of the district.

b – Scheelite-quartz stockwork within the Mine granite, near Signal shaft. Note the regularity of the veins systems.

c – Skarnoid with scheelite within layered Ordovician formations. Scheelite occurs in both stratiform lenses and crosscutting veinlets. Signal mine, level 707.

d – Same skarnoid with a metric fold showing a phlogopite-rich layer. Phlogopite yields a $^{40}\text{Ar}/^{39}\text{Ar}$ age of 285.3 Ma.

e – Facies of the skarn with disseminated sulfides. Visean formation near Kaolin granite

f – Greisen veins and stockwork. This muscovite-quartz stockwork is a continuation of the W1 north vein within Mine granite.

g – Pyrrhotite vein (about 0.8 meter thick) with pyrrhotite-arsenopyrite veinlets at footwall and hanging wall. Signal vein level 707.

h – North side of Mispickel granite (main summit) with Visean (right side). The brownish part between granite and Visean outlines the oxidation zone of the pyrrhotite vein crosscutting the Mispickel granite skarn. Two small galleries have been dug here.

i – Silica breccia from the Tighza structure (photo width: 1m)

(all photographs by E Marcoux, except photograph a by K. Nerci)

Figure 4 - Gold content of the W-type gold veins near the Mine granite (gold analyses by ICP; CMT documents)

Infillings of these veins consist of three stages (Fig. 5):

(1) Stage 1 is an early association of Q_1 quartz with wolframite (crystals of 1 mm to 1 cm) turning to scheelite that reflects a high-temperature oxidizing environment (Figs. 6a, 6b);

(2) Stage 2 is an intermediate stage with gold, bismuth and tellurides, and few sulfides, mostly As-sulfides, coeval with a brittle tectonic event and the Q_2 quartz deposition. Stage 2 is the most abundant in the veins. The near-granite zone (W1 north vein) displays an association of early loellingite and gold, with native bismuth, bismuthinite, tetradymite, joséite B (Bi_4Te_2S), hedleyite (Bi_7Te_3), followed by later minor arsenopyrite (Nerci, 2006) (Figs. 6c to 6h). These mineralogical associations are consistent with the strong statistical correlation between Au and Bi (0.88) and Au and Te (0.95) found among more than one hundred of vein ore samples (Nerci, 2006). These minerals do not occur in the distal zone of the W1-W1 north vein system (W1 vein) displaying a low grade in gold, and more abundant arsenopyrite and chalcopyrite. The strong association of gold with loellingite, and not with arsenopyrite, explains the low As-Au correlation (0.13). Grain (10 à 200 μm) of native gold are associated with bismuth and tetradymite. Electronic microprobe analyses of 18 grains show that gold has homogeneous electrum composition (84% Au, 16% Ag), with regular low Bi values averaging 0.4 %. No gold was detected in pyrrhotite and arsenopyrite by ICP/MS laser analysis (detection level: 1 ppm).

(3) Stage 3 is the final stage with massive pyrrhotite – chalcopyrite and sphalerite (Figs. 5, 7). This stage is mainly observed in drill-holes intersecting W1 north and W4. It is composed of veinlets and veins of massive pyrrhotite, chalcopyrite, with locally abundant sphalerite. Stage 3 form

veinlets cutting earlier grey quartz veins enriched in loellingite-bismuth-gold, or appears as a cement of arsenopyrite–loellingite-native bismuth, or wolframite-bearing quartz clasts (Figs. 7e to 7h).

The “Pyrrhotite vein” strikes N 150° and dips 50-60° NE, with a mean width of 0.8 m. It has been observed in the mine at the 707 and 857 levels and reaches the surface to the south of Signal shaft, where it is marked by strong magnetic anomalies, indicating a vertical extent of at least 580 m (Fig. 3g). This vein is dominated by pyrrhotite, sphalerite and arsenopyrite (Figs. 7c, 7d). At the 857 mine level, both the footwall and hanging wall are silicified and show abundant clasts of arsenopyrite bearing numerous early corroded minerals (loellingite, native bismuth and bismuthinite, no gold). Sn and W are also present. Tin occurs as small ovoid patches (average of 200 µm) of cassiterite in quartz or sphalerite (Fig. 7c), sometimes with a crown of stannite. Tungsten forms interstitial scheelite among sulfides. This “Pyrrhotite vein” is thus dominantly part of stage 3 paragenesis of W veins, and includes relicts of stages 1 and 2 as fragments (Figs. 5, 7h). It is interpreted to belong to the Au-W quartz vein system.

Figure 5 – Paragenetic sequences of the main gold-bearing structures. Most prominent features are illustrated in figures 6 and 7

Figure 6 - Mineralogy of Tighza: ore from early stages

- a – Pseudomorphic crystal of scheelite (Sc) after ferberite (Fb) within quartz (dark grey). A few relicts of ferberite are still preserved. W1 north Au-W quartz vein (sample NK 82 b).
- b – Ferberite (Fb) and loellingite (Lo) within quartz (dark grey). W1 north Au-W quartz vein (sample NK 74).

c – Gold (yellow) and native bismuth (Bi) infilling a quartz crack. Enlargement of the gold particle shows the close association with bismuth (Bi). Drill core on W1 north Au-W quartz vein (DDH SW1N-3 bis).

d – Association of gold (yellow, 14 μm in size) and native bismuth (Bi) within loellingite (Lo). A small patch of pyrrhotite (grey-brown) is also visible. Quartz-scheelite stockwork in the Mispickel granite.

e – Association of gold (yellow), native bismuth (Bi) and bismuthinite (Bs) within loellingite (Lo). Drill core on W1 north Au-W quartz vein (DDH SW1N).

f – Idiomorphic loellingite (Lo) and small patches of gold (circles) within quartz (dark grey). W1 north Au-W quartz vein (sample NK 74)

g – Native bismuth (Bi) with hedleyite (Hd) in quartz (dark grey). Drill core on W1 north Au-W quartz vein (DDH SW1N).

h – Loellingite (Lo) as relicts within arsenopyrite (Apy). Drill core on W1 north Au-W quartz vein (DDH SW1N)

3.2 - Quartz-scheelite stockwork

Quartz-scheelite stockwork corresponds to minor uneconomic occurrences that are developed on the north-eastern edge of the Mispickel granite and within the Mine granite (see Figs. 2, 3b). Three main families of veinlets are distinguished: N170-180°E, N110-120°E and N70-80°E. The mineral association is dominated by wolframite in centimetric crystals, often retromorphosing into scheelite, and loellingite-arsenopyrite. This latter association shows numerous inclusions of native bismuth (up to 300 μm), without bismuthinite, at the contact between the two minerals, and locally with native gold (20 μm).

3.3 - As-Fe-(Au) dissemination in skarn

Significant gold anomalies (up to 5 g/t Au) have been reported in two sites: i) to the southeast of the Kaolin granite, in surface rock samples and in drill-core (S II-4, see Fig. 2); and ii) in drill-cores located north of “Au-W-type” veins near the Mispickel granite (FNJ 4, Fig. 2). At both sites, these anomalous gold-bearing disseminated sulfides are linked to Visean carbonate layers, up to one metre thick, which have been hydrothermally metasomatised into skarn facies. The concentration of the sulfides in specific layers leads to a stratabound habit, systematically associated with crosscutting sub-coeval veinlets, filled with the same minerals.

Near the Kaolin granite, at the surface, these anomalies extend for more than 1 500 m and are characterized by strong grey silicification (up to 10 m thick), with minor arsenopyrite-pyrrhotite (Fig. 3e). They are hosted within a skarn dominated by garnetite (grossular), hedenbergite and wollastonite that is locally replaced by F-bearing vesuvianite (up to 2.2 wt %), epidote and chlorite. Sulfides, more abundant in drill holes, are fine-grained and disseminated, dominated by pyrrhotite and arsenopyrite coming from the destabilization of early loellingite. Accessory minerals are chalcopyrite, loellingite, native bismuth (abundant at the contact loellingite-arsenopyrite), marcasite, primary and secondary pyrite, bismuthinite, sphalerite, and very rare gersdorffite and nickeline (Fig. 7a). Native gold has not been observed. Hydrothermal alteration is dominated by chlorite. Near the Mispickel granite, similar mineralization has been observed in drill-core (FNJ-4) with abundant pyrrhotite, arsenopyrite and pyrite, less bismuth and no loellingite.

In addition, a calc-silicate skarn have been recognized at the contact of Visean limestones with the Mispickel granite (called “skarn of Mispickel granite” on Fig. 2). The mineralogical association shows four successive stages of calc-silicates: 1) abundant green zoned andradite (An₅₉-Gr₃₈) with

small amounts of tin (0.12% SnO_2), vesuvianite, wollastonite and diopside; 2) hedenbergite, 3) hydrogrossular and 4) a retrograde evolution giving an actinolite, chlorite and epidote assemblage.

This calc-silicate skarn itself is barren, but it is crosscut by a sulfide-rich vein that contained massive pyrrhotite, chalcopyrite and cubanite (Fig. 2, 3h, 7b). These minerals strongly suggest that the vein is of the same type as the “Pyrrhotite vein” and thus belongs to stage 3 of Au-W quartz veins. Pyrrhotite locally shows abundant small inclusions of early loellingite, native bismuth and arsenopyrite, with some gold (equivalent to stage 2 of the veins). Anomalous gold grades were measured in a very small number of samples from the skarn (3 out of more than 200 samples) and reflect the presence of quartz-loellingite-gold bearing-clasts, indicating a tectonic brecciation of the early W-Au quartz veins with the angular clasts having been reworked and cemented by the late pyrrhotite of stage 3.

The pyrrhotite-arsenopyrite-sphalerite assemblage of this As-Fe-(Au) association is fairly similar to stages 2 and 3 observed in the Au-W quartz veins and could thus represent the end of gold deposition process.

Figure 7 – Mineralogy of Tighza: ore from late stages

a – As-Fe-(Au) dissemination in skarn showing early arsenopyrite crystals (Apy) corroded by pyrrhotite (Po) within quartz. Noticeable gold grades (>5 g/t) have been measured in this facies.

South-East of Kaolin granite (drill-core S II 138 m, sample NK 116).

b – Massive pyrrhotite (Po, grey-brown) with chalcopyrite (Cpy) and residual arsenopyrite (Apy).

Pyrrhotite-sphalerite-(chalcopyrite) vein crosscutting the Mispickel granite skarn.

c – Mineralogy of the pyrrhotite vein. Dominant pyrrhotite (Po) and sphalerite (Sp) are accompanied by chalcopyrite exsolutions (Cpy) and rare cassiterite (Cs). Signal mine, level 707, sample NK 109.

d - Mineralogy of the pyrrhotite vein. Pyrrhotite (Po) and sphalerite (Sp) show residual patches of arsenopyrite-loellingite (Apy). Black patches are silicates. Signal mine, level 707, sample NK 109.

e – Clast of arsenopyrite-loellingite (Apy) and quartz (black) (stage 2) cemented by later pyrrhotite (Po) and chalcopyrite (Cpy) within a secondary quartz (stage 3). The pyrrhotite - chalcopyrite association is similar to that of pyrrhotite vein (see photo d). Drill-core cutting W4 vein at 93 m DDH SW4J-2.

f - Clast of arsenopyrite-loellingite (Apy) with bismuth inclusions (Bi) within quartz (black) invaded by later pyrrhotite (Po)-chalcopyrite (Cpy), within quartz (black). The pyrrhotite - chalcopyrite association is similar to that of pyrrhotite vein (see photo d). Drill-cores cutting W4 vein at 93 m (photo e, DDH SW4J-2) and 160 m (photo f, DDH SW4J-4).

g – Quartz clast with loellingite (Lo) with gold (Au) and bismuth (Bi) inclusions (stage 2) wrapped in pyrrhotite (Po) – chalcopyrite (Cpy) (stage 3). An enlargement of the gold-bismuth patches is shown. Occasional gold grades in rare bulk pyrrhotite-bearing samples is due to such gold-bearing clasts. Drill-core cutting W4 vein at 160 m (DDH SW4J-4).

h – Quartz clast with elongated crystals of ferberite (Fb) largely transformed into scheelite (Sc) (stage 1) and residual arsenopyrite (Apy; stage 2), wrapped into pyrrhotite (Po; stage 3). Quartz is very dark grey. Pyrrhotite vein, Signal mine, level 707, sample NK 109. This photograph shows that early oxidized paragenesis of stage 1 can also be present as clast in this vein

3.4 - Mineral compositions

Electron microprobe analyses were performed on several minerals (arsenopyrite, loellingite, sphalerite, and pyrrhotite) associated with the various types of Au mineralization in the Tighza area (Table 1). Scheelite were also analysed (Table 1). These analyses allow to check the similarity of traces elements in mineral from different locations.

Table 1 – Average composition of sulfide minerals. Number of analyses is within brackets. Electron microprobe analyses EPMA-Electron Probe Micro-Analysis) CAMECA SX 50 (BRGM-ISTO, Orléans, France). Tension acceleration: 20 kV, current: 10 nA. Co, Ge, Ag and Ni have not been detected in sphalerite. Sb is under detection limit in arsenopyrite

Arsenopyrite exhibits a wide range of compositions, even at the vein scale (Fig. 8). The arsenopyrite compositions of various ore types widely overlap: 31.53 to 34.26 (average 32.73 at.% As) in W1 vein, 33.0 to 36.5 (average 34.95 at. % As) for W1 north vein, 33.4 to 37.0 at.% As for the pyrrhotite vein. In the As-Fe-(Au) disseminations within skarns, stratabound arsenopyrite has a composition (31.5 to 34.6 at.% As) slightly different from its associated crosscutting arsenopyrite (31.58 to 32.45 at.% As). Trace elements are more significant. Co is present in low but systematic amounts in W1 veins (up to 0.89 wt % Co in W1 vein, and up to 1.55 wt % in W1 north vein) especially at the surface level, and in the pyrrhotite vein (0.10 to 0.20 wt % Co). Trace elements in arsenopyrite from the As-Fe-(Au) disseminations are different: Co is absent, but Bi and Hg are always present in low contents (up to 0.44 wt %) in stratabound arsenopyrite. Antimony has not been detected.

Figure 8 - Bivariate plot of the Co versus As content of arsenopyrite (see Table 1).

Loellingite has a homogeneous composition with constant sulfur content (average 1.65 wt % S) and occasional Co and Ni contents (about 1 wt % Co+Ni). As-Fe-(Au) disseminations in the skarn of Kaolin granite are sulfur-poor (0.5 wt % S) with cobalt and nickel-rich loellingite (4.0 wt % and 9.6 wt % respectively). Pyrrhotite has been analyzed in the pyrrhotite vein of the Mispickel granite skarn, the pyrrhotite vein of the mine, the Au-W vein and the As-Fe-(Au) dissemination in the skarn of Kaolin granite. The pyrrhotite compositions are very constant, ranging from 38.5 and 40.0 wt % S, implying formulas ranging from $\text{Fe}_{6.67}\text{S}_8$ to $\text{Fe}_{7.02}\text{S}_8$. Sphalerite analyses were limited to the pyrrhotite vein and to the W4 vein drill-cores. Compositions are similar at both sites and exhibits a fairly high iron content (5.8 to 12 wt % Fe, giving 8 to 22.5 wt % FeS), low Cd content (average 0.28 wt %), but no silver. Scheelites from different ore facies (stockwork, skarn, and level 707 W-Mo vein) were also analyzed: Pb and Mo contents are too low (respectively 0.37 and 0.32 wt %) to distinguish different populations of scheelite.

4 - Fluid inclusions

4.1 - W-Au system

The main tungsten and gold-bearing veins were studied for fluid inclusions in quartz Q_1 and Q_2 (Nerci, 2006, see annex for methodology). All gold and tungsten bearing-quartz exhibits a two-phase infilling with an earlier hyaline quartz Q_1 and a later saccharoidal quartz Q_2 . As mentioned above, Q_2 derived in part from Q_1 by microfracturing but also comprise neo-formed crystals resulting from silica-rich hydrothermal flows. Two types of fluid inclusions have been identified in Au-W-type veins and within the quartz-scheelite stockworks of the Mine and Mispickel granites:

i) low-salinity aqua-carbonic inclusions (type IF1); ii) high-salinity inclusions with at least one solid phase (type IF2) (Fig. 9).

The IF1 type corresponds to sparse pseudo-secondary aqua-carbonic inclusions, only observed within the Q_1 quartz from the W-Au quartz veins. IF1 ranges between 5 and 40 μm , and are frequently three-phased ($L_{\text{H}_2\text{O}}+L_{\text{CO}_2}+V_{\text{CO}_2}$, Fig. 9A). Generally, by visual estimate, IF1 contain >50% vol. of aqueous liquid with respect to CO_2 phases. Large portions of these inclusions are exploded or decrepitated and present various morphologies (e.g. lobate, dendritic, half ring....). Very few solid are observable within this inclusion type. The hyaline Q_1 quartz being well preserved in the W1 north vein, hereafter are presented mean and range values from 15 studied inclusions. Only six accurate T_h measures were possible yielding a 240°-310°C range with homogenization in vapor. $T_{m\text{CO}_2}$ ranges between -57.1°C and -56.7°C, whereas $T_{h\text{CO}_2}$ ranges between 22.5°C and 27.5°C in vapor. These temperatures have been reported in the V-X projection of Thiéry et al. (1994) for the CO_2 - CH_4 system: X_{CH_4} is about 0.5-3.2% mol and the molar volume of volatile phase ranges between 143 and 169 cm^3/mol . Raman spectroscopy confirmed the presence of CH_4 . Although $T_{m\text{ice}}$ is difficult to determine for each inclusion, we obtain a mean value of -3.7°C on three inclusions. Assuming a CO_2 contribution of -1.48°C (ice melting temperature in the H_2O - CO_2 system), the NaCl contribution is estimated around -2.22°C. This yields a salinity of 3.7 wt % NaCl from H_2O -NaCl equiv. system equation (Bodnar, 1993). Molar fractions and global density have been estimated with the method developed by Ramboz et al. (1985): compositions are homogenous with 1 mol % NaCl, 17-19 mol % CO_2 , 80-82 mol % H_2O and traces of CH_4 , for a global density of $0.53\pm0.1 \text{ g/cm}^3$. Finally, using isochores from the equation of Bowers and Helgeson (1986) for the CO_2 - H_2O -NaCl system, the trapping temperature and pressure were evaluated in ranges of 400-600°C and 640-1490 bars. Because the Q_1 quartz hosts the W-bearing minerals, the IF1 type, pseudo-secondary inclusions, are considered to be

associated with the early wolframite-scheelite stage. On the scale of the district, the IF1 inclusions are characterized by deformation, re-equilibration and explosions, coeval with a cataclasis of hyaline quartz Q_1 , and appearance of “dry” microcracks, without fluid inclusion.

Figure 9 – Fluid inclusions types and isochores obtained for the second generation of fluid inclusions (IF2) in the Au-W (A, B, C and D) and for the Pb-Ag mineralizing systems (isochores only). A: typical biphased inclusion with liquid and vapor phases; B: fluid inclusion planes within Q_1 ; C and D: high-salinity inclusions with hyalite cube (C) and hyalite cubes and dark inclusion of chalcopyrite. All isochores are constructed in the H_2O -NaCl system. Lithostatic and hydrostatic gradient were constructed using a reasonable geothermal gradient of $50^\circ C$ per km with mean densities of 2.5 and 1 for rocks and water-rich fluids respectively. For IF2 fluids trapping PT conditions are bracketed between 200 Mpa/ $400^\circ C$ in lithostatic regime and 42 Mpa/ $267^\circ C$ in hydrostatic regime. This very large range is discussed in further section in text. For mineralizing brines of the Pb-Ag systems, textures diagnostic of hydrostatic conditions favor trapping conditions at 27Mpa/ $242^\circ C$.

The IF2 type is found: i) within Q_1 along numerous fluid inclusion planes (12 studied IF2, Fig. 9B); ii) as primary inclusions within Q_2 neo-grains (Fig. 9C); iii) within fractured quartz of quartz-scheelite stockworks (20 studied IF2, Fig. 9D). They clearly post-date IF1. IF2 are multiphased inclusions with liquid and vapor phases (max 15% vol. ratio) still in the presence of solids (max 30% vol. ratio). Halite crystals are easily recognizable by their cubic geometry and refringence. Raman spectroscopy allows us to identify refringent solids such as carbonates (calcite), micas, and chalcopyrite (Fig. 9D), T_{SNaCl} (dissolution temperature) ranges between $220.5^\circ C$ and $282.2^\circ C$. These dissolution temperatures yield salinity range of 33.65-36.80 wt % NaCl equiv. in the H_2O -

NaCl system (Bodnar and Vityk, 1994). Homogenization temperatures (gas bubble resorption) give a 257-284°C range with $T_h > T_{SNaCl}$. Because Q₂ quartz is associated with the gold-bearing stage 2 (see above), it is likely that these IF2 fluids were rich in Au-Te-Bi-Cu (and possibly Si).

4.2 - Pb-Ag-Zn system

In order to facilitate a comparison with the W-Au deposition system, a limited study was carried out on fluid inclusions within calcite associated with the Pb-Ag-Zn mineralization in the Sidi Ahmed vein (15 IF measurements). The banded and geodic textures of quartz and carbonates suggest a shallow deposition process under hydrostatic fluid pressures. Most of the inclusions are liquid-dominated and show a gas and a solid phase (carbonate or salt cube). Two families of fluid could be distinguished based on T_e at -50°C and -70°C (secondary inclusions). The first one reflect Na-Ca enriched brines whereas the complex melting and freezing behavior (irreversible) of the second suggest presence complex fluid system, implying organic compounds for instance. T_{SNaCl} determined for the first type presenting salt cubes, yield salinities around 31 wt % NaCl equiv. in the H₂O-NaCl system (Bodnar and Vityk, 1994). Homogenization temperatures are rather dispersed between 165° and 217°C for the primary generation and around 180°C for the second. Both homogenize in liquid phase.

4.3 - Trapping conditions

Several isochores have been traced using the Loner 32 program (Bakker, 2003) in the H₂O-NaCl system (Knight and Bodnar, 1989; Bodnar, 1993; Bodnar and Vityk, 1994) (Fig. 9E). To estimate a true trapping temperature, a pressure correction was applied to the homogenization temperatures based on a mean thermobarometric gradient of 50°C and 25 Mpa per km for the district when

granitic magmatism and hydrothermal events occurred. For the W-Au system, IF2 type fluids were trapped at maximum conditions near 200 Mpa and 400°C (Fig. 9E: intersection between the lithostatic gradient and the isochore calculated from the maximum Th) and minimum conditions near 40 Mpa and 265°C. In the Pb-Ag-Zn system, accounting for veins textures which are diagnostic of large permeabilities and hydrostatic conditions, minimal trapping conditions are estimated around 30 Mpa and 250°C (Fig. 9E).

5 - Isotopic geochemistry

$^{40}\text{Ar}/^{39}\text{Ar}$ dating was done to better constrain the age of magmatic and hydrothermal events. Pb isotope analyses were performed in order to constrain the origin and relationship of the various types of ores, and the possible link between Pb-Ag-Zn veins and granites. Similarly, Sm-Nd and Sr isotopes of scheelite from various ore types were performed to constraint the relationship of the different ore systems.

5.1 - $^{40}\text{Ar}/^{39}\text{Ar}$

Two biotites/phlogopites from the Mine granite and W1 vein and two muscovites were dated using a $^{40}\text{Ar}-^{39}\text{Ar}$ laser probe (see annexes for methodology). The Mine granite biotite (NK 211) yielded an age of 286.0 ± 0.4 Ma, whereas the W-skarn (NK 222) and W-Mo vein (NK 220) from level 707 in Signal mine dated using muscovite yielded 285.3 ± 0.5 Ma and 285.6 ± 0.5 Ma respectively (all uncertainties: 1 σ level) (Table 2 and Fig. 10).

These new data are consistent with and provide further constraints to the previous K-Ar dating by Cheilletz (1984) who dated the Mine granite at 287.9 ± 5.1 Ma and 291.1 ± 15.2 Ma, the W-skarn

at 285.3 ± 3.0 Ma and 284.1 ± 4.2 Ma, and the W-vein at 288.4 ± 3.1 Ma. The W-Au mineralization is therefore strictly coeval with the cooling of the Permian-aged Mine granite.

It is noteworthy that a fresh phlogopite sampled at the surface on the W1 north vein (NK 313) gave an age of 291.8 ± 0.3 Ma, slightly older than the above ages, older than the K-Ar age obtained on the W1 by Cheilletz (1984) at 285.0 ± 3.8 Ma, and also older than the Mine granite. This apparent discrepancy will be discussed later (see Discussion). Dating of associated muscovite (NK 312) did not yield a viable spectrum.

Table 2 – $^{40}\text{Ar}/^{39}\text{Ar}$ analytical data. $^{40}\text{Ar}_{\text{atm}}$ = atmospheric ^{40}Ar . $^{40}\text{Ar}^*$ = radiogenic ^{40}Ar . Ca = produced by Ca-neutron interferences. K = produced by K-neutron interferences. Age (Ma) = the date is calculated using the decay constants recommended by Steiger and Jäger (1977). The errors are at the 1σ level and do not include the error in the value of the J parameter. Correction factors for interfering isotopes produced by neutron irradiation in the McMaster reactor were $(^{39}\text{Ar}/^{37}\text{Ar})\text{Ca} = 7.06 \times 10^{-4}$, $(^{36}\text{Ar}/^{37}\text{Ar})\text{Ca} = 2.79 \times 10^{-4}$, $(^{40}\text{Ar}/^{39}\text{Ar})\text{K} = 2.97 \times 10^{-2}$

Figure 10 - ^{39}Ar - ^{40}Ar age spectra. The age error bars for each temperature steps are at the 1σ level and do not include errors in the J-values. The errors in the J-values are included in the plateau age calculations

5.2 - Lead

Twenty-one sulfide samples, mainly galena, were collected from the different mineralized structures, and two K-feldspars from the Mine and the Mispickel granites have been analyzed. These data are complemented by Pb isotope data from the Signal Pb-Ag-Zn vein (Watanabe,

2002). The lead isotope data are widely dispersed, ranging from 18.25 to 18.90 ($^{206}\text{Pb}/^{204}\text{Pb}$ ratio), suggesting the interaction of several sources (Table 3 and Fig. 11)

The $^{206}\text{Pb}/^{204}\text{Pb}$ isotope ratios of galena from gold-bearing veins (W-Au quartz veins) define a fairly wide range from 18.29 to 18.42 and are independent of vein location or proximity to the Mine granite. The isotopic compositions of the pyrrhotite vein and of the Mispickel pyrrhotite deposit overlap the least radiogenic $^{206}\text{Pb}/^{204}\text{Pb}$ of the W veins, although As-Fe-(Au) skarn compositions fall within field for the W-Au quartz veins (18.299 to 18.343 for $^{206}\text{Pb}/^{204}\text{Pb}$ ratio).

There is no distinction between northern (Mispickel granite) and southern (Kaolin granite) skarns. Nevertheless, the Pb isotope compositions of sulfides (pyrrhotite+sphalerite) of the W-skarn associated with the Mine granite yield more highly radiogenic and dispersed ratios (18.27 to 18.90 $^{206}\text{Pb}/^{204}\text{Pb}$ ratios) and are reminiscent of sulfides associated with black shales or limestones. The K-feldspar compositions are close to 18.30 ($^{206}\text{Pb}/^{204}\text{Pb}$ ratio) and are similar to values of Hercynian granites (Marcoux, 1987). Thus these granites are not likely a major source for the W-Au mineralization.

The lead isotopic compositions of the Pb-Ag-Zn major veins are clustered in a relatively narrow field (18.265 ± 0.035 for $^{206}\text{Pb}/^{204}\text{Pb}$ ratio) and overlap the lowest $^{206}\text{Pb}/^{204}\text{Pb}$ ratios of the W-Au mineralization (Fig. 11). The new analyses performed for this study are identical ($\pm 3 \sigma$) to those published by Watanabe (2002).

The results show what can be considered as a short fractionation line with Sidi Ahmed Pb-Ag-Zn veins forming the least radiogenic end-member at 18.232 ($^{206}\text{Pb}/^{204}\text{Pb}$) and 15.68 ($^{207}\text{Pb}/^{204}\text{Pb}$). Although these veins are the youngest in the district, they exhibit the least radiogenic ratios.

As a whole, the lead isotopic compositions of Tighza district are characterized by high $^{207}\text{Pb}/^{204}\text{Pb}$ ratios, indicating a source in the upper continental crust (Stacey and Kramers, 1975; Doe and Zartman, 1979).

Figure 11 – Pb-Pb diagram of the various type of mineralization. Stacey-Kramers (1975) growth curve and Doe and Zartman (1979) orogen curve are mentioned (see Table 3)

Table 3 - Lead isotope compositions of Tighza ores. Samples analyzed Ag-Zn veins (galenas) and K-feldspars. Analyses performed by J.J. Peucat (University of Rennes France), except (1) by Watanabe (2002). Analytical uncertainties see annexes

5.3 - Sr and Sm/Nd

Sr and Nd isotope data for scheelite and whole-rock samples from the Tighza area are presented in Table 4 and Figure 12. The Sr isotope compositions of the scheelites are considered initial ratios due to the very Sr rich (>1000 ppm) and Rb-poor (<5 ppm) nature of the mineral. The analyses of the scheelites yield Sr isotope compositions that vary from 0.7089 to 0.7136. These results are, for the most part, less radiogenic and more variable than the Sr isotope data obtained from siderite of Pb-Ag veins analyses by Castorina and Masi (2000). The greater variability obtained in this study likely reflects the more widespread sampling of this study and the less radiogenic Sr isotope ratios are evidence of the contribution of a less radiogenic fluid component. The variation in the scheelite Sr isotope ratios at Tighza is also independent of sample localities in the granite and the overall vein structures/types. For example, two different, but closely spaced gold veins (W4 and W5)

yield different Sr isotope ratios of 0.710 and 0.713, respectively. Although the Tighza scheelites partially overlap the initial Sr isotope compositions of the Mine Granite, Mispickel granite and granites from Castorina and Masi (2000), the granites are generally exhibit the least radiogenic (most primitive) compositions. Nd isotope analyses of the granitic and rhyolitic rocks yield typical crustal Nd contents (21-31 ppm) and $^{147}\text{Sm}/^{144}\text{Nd}$ ratios (0.096 – 0.122). The scheelites are much richer in Nd (215 - 636 ppm) and have near chondritic $^{147}\text{Sm}/^{144}\text{Nd}$ ratios (0.186 – 0.218). The initial ϵNd values (at 300 Ma) of the granites and rhyolite range from -3.6 to -6.6 and are more radiogenic (and thus more primitive) than the range of initial ϵNd values (-2.4 to -8.2) found in the scheelites.

Table 4 – Sr and Nd results (analyses: R. Stevenson, GEOTOP/UQAM Montréal, Canada)

Figure 11 - Plots of ϵNd versus $87/86\text{Sr}$ (a) and $87/86\text{Sr}$ versus Sr concentration (b) for lithologies and mineral deposits of the Tighza district. Field for Cambrian basement (sedimentary rocks) from Schaltegger et al. (1994), and additional granite and siderite data from Castorina and Masi (2000). The Sr and Nd isotope data argue for an interaction between basement sedimentary fluids and primitive granite-derived fluids

6 - Genesis of the Tighza polymetallic district

Until now, each of the different styles of mineralization has been interpreted separately (Cheilletz and Giuliani, 1988; Castorina and Masi, 2000, Watanabe, 2002). The above data allows us to propose an integrated model for the emplacement of the successive mineralizing events in the Tighza area.

6.1 - Plutonic events and igneous metallogeny

The 286.0 ± 0.4 Ma ^{40}Ar - ^{39}Ar age obtained on biotite from the Mine granite confirms the previous published ages (Cheilletz, 1984). The emplacement of the granitic stocks is strongly controlled by wrench faulting. Particularly, the Mine granite, which corresponds to an apex of a larger underlying plutonic complex (Eldursi et al., 2009), is emplaced along a dilational horsetail termination of the dextral W-veins system, during the D2 tectonic event. Such setting indicates that the structural level corresponds to the brittle/ductile transition in high geothermal regime.

The geochemistry of the three Permian granitic stocks is relatively homogeneous (Table 1). The degree of fractionation with respect to the degree of oxidation of the parent magmas is shown on the Rb/Sr vs. $\text{Fe}_2\text{O}_3/\text{FeO}$ plot for the three main stocks and is clearly located on the boundary of the IRGD domain with the W domain (Fig. 13). The parent granitic magma is moderately oxidized and fractionated close to the clan of the “Intrusion-related gold deposits” associated with a W-Mo metal assemblage, as defined by Blevin et al. (1996) and Blevin (2004). Such a moderate state of oxidation and fractionation may be interpreted as a mantle origin of the magma partly reduced by the abundant crustal organic matter within the Paleozoic rocks.

Figure 12 – Oxidation and fractionation state of the three main Permian stocks of the Tighza area (black square: Mine granite; white square: Kaolin granite; gray square: Mispickel granite). Oxidation and fractionation of magmas associated with different dominant metal assemblage are schematically represented. Area limited by red dash line corresponds to the magma composition from the Intrusion-related Gold Deposits (IRGD) (from Blevin et al., 1996, and Blevin, 2004)

6.2 - W-Au events

Three paragenetic events can be distinguished among the Tighza W-Au occurrences.

Stage 1 consists of an early association of wolframite followed by scheelite, represented by early stratabound lenses of skarns and W (not yet Au) quartz veins; the early W-only association reflects clear sub-magmatic conditions typical of a reduced skarn environment (Einaudi et al., 1981). Initial temperatures of the hydrothermal system reach 580°C and 1.8 Kb during early skarn mineralization and 400° to 450°C and 100 MPa for vein-style deposits (Cheilletz and Isnard, 1985). The IF1 CO₂-rich fluids are interpreted to be coeval with stage 1 (Fig. 9). These CO₂-rich fluids are occluded as primary or pseudo-secondary inclusions within Q₁ quartz which is coeval with W minerals. Therefore, the spatial link of W mineralization with the Mine granite (i.e. the inner W circle) strongly suggests that CO₂ was exsolved from felsic melts as demonstrated in numerous modern systems (Etiope and Martinelli, 2002). Exsolution of early CO₂-rich fluids from felsic melts is well established as CO₂ is much less soluble than H₂O in melts (e.g. Fogel and Rutherford, 1990).

Stage 2 consists of an intermediary gold-bearing stage with native gold, bismuth and tellurides, and a few sulfides, mostly loellingite and arsenopyrite; W-rich quartz veins reflect the transition from a pervasive metasomatic process with a limited permeability to circulation in connected fractures, therefore implying a transition from an aseismic to a seismic environment. The zoning in the W1 vein from the Au-rich to the As-rich zones could be the result of both temperature and fugacity, where the center of the hydrothermal system is located on the Mine granite with high temperature, and fS₂ increase outward in the shale-hosted host-rocks. The temperature of deposition of this gold stage can be deduced from the paragenetic association, specifically the presence of rare Bi-Te minerals. According to Afifi et al. (1988) and Dimitrova and Kerestedjian (2006), the bismuth+hedleyite association exists only below 266°C. A minimum temperature for

presence of pyrrhotite and bismuthinite is 235°C (Afifi et al., 1988). With a decreasing temperature, loellingite development and disappearance of bismuthinite correspond to a decrease in the S_2 fugacity (Fig. 14). The presence of Te minerals is probably an indication of the condensation of evolving magmatic fluids as Te has a low solubility in this type of fluids (Cook and McPhail, 2001; Tombros et al., 2010). The presence of nickeline could reflect enhanced Ni mobility in sulfur poor systems under approximately neutral to acidic conditions (Bermanec et al., 2000). These results confirm the measured T_h values for fluid inclusions (IF2), between 257 to 284°C. A pressure of 20 MPa for a hypersaline fluid can be determined from the NaCl fluid stability diagram (Liebscher, 2007). As mentioned above the trapping pressure for the IF2 inclusions is difficult to pinpoint (Fig. 9E), homogenization being achieved by bubble disappearance rather than by halite dissolution ($T_h > T_{S_{NaCl}}$). However, a low trapping pressure for IF2 fluids is also consistent with the small differences in temperatures determined from mineral. This would imply therefore a large decrease in pressure from the previous W stage. Such decrease may have been facilitated by a transition from lithostatic to hydrostatic pressure. Indirect spatial and temporal constraints on IF2 trapping conditions will be implied and discussed in the next section.

Stage 3 consists of massive pyrrhotite – chalcopyrite and sphalerite, and accessory Sn minerals (cassiterite, stannite). The third stage appears after a fracturing event, and either cements quartz-gold-sulfides clasts of the previous stages (W veins), or more often forms independent deposits (the Pyrrhotite vein and sulfide vein that crosscut the skarn of the Mispickel granite). Late Zn-Cu-(Sn) stage 3 deposition of massive pyrrhotite – chalcopyrite – sphalerite is characterized by lower sulfur fugacity ($\log f_{S_2} < -11$) as indicated by the composition of the sphalerite (8-22.5 at.% Fe) coeval with pyrrhotite crystallization.

The Tighza W-Au system is pyrite-depleted, pyrrhotite-rich, with numerous As-sulfides that, when combined with the presence of Bi-Te minerals and the low homogenization temperature, demonstrate that the system as a whole is sulfur-deficient.

Figure 13 – Evolution of the conditions of deposition in a fS_2 - t °C diagram (grey arrow), established from fluid inclusion and paragenetic studies. Scheelite-skarn temperature formation (580°C, Cheilletz and Isnard, 1985) has been considered as the starting conditions. Stage 1 conditions are deduced from CO₂-rich IF1 fluid inclusion studies, paragenetic assemblages and As at.% of arsenopyrite. Stage 2 conditions (265-400°C) take into account paragenetic equilibrium observed and IF2 fluid inclusions. Stage 3 (<265°C) drastical change in fS_2 is a result of sphalerite iron content. Deposition conditions of late Pb-Zn veins are not figured. Equilibrium curves and stability fields are from Kretschmar and Scott, 1976, Vaughan and Craig, 1997, and Barnes, 1997

The spatial setting of these three stages defines a perigranitic zoning model, with an inner W circle, a middle circle up to 400 m with Bi, Te, Au and As. From the granite border outward, loellingite is replaced by arsenopyrite and gold grade shows a clear decrease from 21 ppm within the 200 m closest to the granite to 5 ppm over the next 500 m (see Fig. 4). The most economic gold target, the Au-W quartz veins, displays these three paragenetic stages with stages 1 and 2 being the most representative.

6.3 - Pb-Ag-Zn events

Pb-Ag-Zn veins record a different hydrothermal event compared to that of the W-Au quartz veins. The structural style is clearly associated with the regional left-lateral wrenching D3 event. Fluid inclusion microthermometry suggests a minimum temperature around 230°C in the Sidi Ahmed

Pb-Ag-Zn vein and geodic textures and collapse breccias suggest emplacement at a shallow level in a low hydrostatic pressures environment.

6.4 - Fluids and metals sources

The radiogenic isotope studies (Pb, Sr, Nd) indicate that there is a strong link between the different mineralization stages and magmatic/crustal fluids. The initial Sr and Nd isotope ratios associated with the early W mineralization overlap the field of isotopic compositions from the Cambrian shales/schists of the Moroccan basement (Schaltegger et al., 1994), and extend towards the primitive Sr and Nd isotope ratios of defined by the Hercynian granites. This suggests a convective cell involving fluids from the granites (Mine and Mispikel) and fluids from the enclosing rocks (Cambrian basement) led to the deposition of the early W (and gold) mineralization. This model can explain the presence of a more crustal fluid in the formation of the siderites (Castorina and Masi, 2000) and is consistent with the model of hydrothermal W deposits of Heinrich (2007). This strongly suggests that the fluids responsible for the formation of Tighza W-Au mineralization were a mixture of fluids remobilized from the underlying Cambrian basement and fluids derived from juvenile granitic magmas.

The compositional field of the different ore types can be interpreted representing mixing between a less radiogenic component (the local granite intrusions) and a more radiogenic component often found in black shales and carbonated rocks.

A plot of the lead isotope data on usual Pb diagram suggests that the different mineralization events were linked to two different hydrothermal systems (Fig. 11). The earlier mainly advective system was associated with the vein type W-Au mineralization, while the second convective system was associated with the later Pb-Ag-Zn veins mineralization. Both advective and

convective systems appear to share a common magmatic component as defined by the K-feldspars from the local granites, but diverge at higher $^{206}\text{Pb}/^{207}\text{Pb}$ ratios towards the values defined by the W-skarnoids (Fig. 15).

7 - Discussion

7.1 - An integrated structural-hydrothermal model

With the exception of the Sb-Ba veins that were not studied in this paper, the Tighza polymetallic district appears to be the result of the superposition of two major metallogenic events (Fig. 15):

- 1) An early W-Au-(Cu-Zn-Mo) mineralization event related to a late-Hercynian (286 Ma) magmatic event and;
- 2) A late event with Pb-Ag-Zn veins that are independent of, and crosscut granite and previous W-Au mineralization.

The location of the gold system is restricted to the proximity of three small granite apexes (Mispickel, Mine and Kaolin granites). The termination of the W1 north vein at the contact of the Mine granite and its strike extension along a hydrothermalized faulted corridor with greisen alteration (see Fig. 2 and 3f) strongly suggests that it predates granite emplacement (Fig. 15A). This interpretation is supported by the younger age of the granite (286 Ma), relative to the phlogopite of W1 north vein ($291.8 \pm 0.3\text{Ma}$, Table 2 and Fig. 10). There is clear evidence that the W-structures were active during the three stages of the W-Au mineralizing event (e.g. the fracturing and brecciation between the different stages). This occurred during the D2 right-lateral transtensive tectonic event. Furthermore, structural patterns and chronological constraints strongly

argue for emplacement of the Mine granite stock within the transtensional zone constituted by the horsetail termination (W1 north, WO3, W4, W5, W6, see Fig. 2) of the W-structures during the D2 right-lateral wrenching (Fig 15B). The structural context indicates that this W-Au was related to a compressive D2 stress pattern, with the main principal stress oriented NW-SE. Therefore, the intrusive process that generate an Herzian concentric stress pattern with a vertical main principal stress during granite emplacement was already ended.

Movements associated with the formation of the “greisen” corridor resulted in the crushing of Q_1 wolframite-quartz (stage I), generating a more saccharoidal, porous quartz (Q_2), that was more favorable to a diffusion of gold and sulfides-bearing fluids (stage 2). This supports the hypothesis of the release of magmatic gold-bearing fluids during the late-crystallization stage of the magma in pre-existing E-W corridors, including the major W1 and W1 north veins, and their WO3, W4 and W5 associated structures (Fig. 15B). These corridors were thus reactivated and invaded by fluids as long as there was magmatic activity in the district, i.e. at least for 5 Ma (291.8 Ma to 285 Ma). A strong mineralogical and geochemical zoning is apparent around the pluton, with a decreasing gold content and an increasing As association (Fig. 4, 15C). Such zoning was a result of a thermal gradient during gold-tungsten deposition.

The variation in the depth of the mineralizing system between stage I (W) and 2 (Au bearing) is difficult to estimate because IF2 fluid inclusions do not yield unequivocal minimum trapping pressures. The numerous exploded inclusions observed among the IF1 inclusions argue for rapid depressurization of the system. However, the concentric geochemical and mineralogical zonation between the W and Au stages around the Mine granite, the strong genetic relationships between granites and W-Au event, the lack of a telescoping intrusion and the short time span for the emplacement of granitic stocks of similar composition in general (no more than 100 000 years, e.g.

Petford et al., 2000) strongly argue for deposition of W and Au stage (and probably the pyrrhotite bearing stage 3 too) at approximately the same depth. Explosion and decrepitation of IF1 inclusions might be then result either from late large-scale rapid uplift after the W-Au mineralizing event (e.g. during the Pb-Zn-Ag event), or from rapid decompression caused by severe D2 incremental fracturing along the W-structures between the W and the Au stages. This last hypothesis is favored because it fits well with the strike slip movements observed in Pb-Zn-Ag veins and the absence of exploded or decrepitation features in the IF2 inclusions. This would suggest a transition from lithostatic to hydrostatic fluid pressure between W stage and Au stage. Thereafter, if CO₂-rich fluids (IF1) were trapped around 1.5 kbar during the lithostatic regime (i.e. depth of 5.5 km, Cheilletz and Isnard, 1985), the minimum trapping pressure for saline H₂O-rich fluids (IF2) would be the hydrostatic value at this depth, i.e. around 550 bars. Such transition from lithostatic to hydrostatic fluid pressure regime has been frequently documented in many non-porphyry intrusion-related ore deposits (e.g. the skarn-related iron deposits of Vegas Peladas; Pons and Franchini, 2009).

Figure 14 - 3D and polyphased genetic model for Tighza

Stage 1 – Incipient right-lateral transtensive D2 event, responsible for the formation of premagmatic “W” structures. Primary hydrothermal fluids (dash line arrows) have likely percolated along those faults and are responsible for phlogopite deposition around 291 Ma. Mineral phases associated with phlogopite have not been recognized in this study.

Stage 2 – Global Au-W stage (ca 285 Ma). Veins emplaced under pressures of 1.5 kbar, i.e. at 5-6 km depth. On the right, upper drawing shows emplacement detail during early W-stage, sub-coeval of intrusion, lower drawing detail during cooling, coeval of stage 2 (gold deposition) and stage 3 (pyrrhotite deposition).

Stage 3 – Pb-Zn-Ag stage following uplift at ca 265 Ma. This stage occurred at least after tonalite intrusion, under 200 bars at hydrostatic pressure

Because IF2 fluid inclusions clearly postdate the IF1 inclusions, they do not represent co-existing fluids and therefore do not suggest immiscibility or boiling during the W-Au event at Tighza. Increasing salinities from IF1 to IF2 inclusions do not favor mixing and dilution by meteoric fluids. The succession from CO₂-rich IF1 fluid to H₂O rich IF2 fluids could be explain by two models:

1) CO₂-rich IF1 fluid represents metamorphic fluid associated to contact metamorphism and decarbonation of the calcareous rocks; H₂O-rich moderate to high salinity IF2 fluids would represent magmatic fluids associated to the Au-W mineralizing events; However, it seems unlikely that CO₂-rich fluids, liberated by decarbonation reactions during contact metamorphism would precede exsolved magmatic brines from melt.

(2) CO₂-rich and H₂O-rich fluids represent the sequential release of an evolving magmatic fluid: as CO₂ solubility in silicate melts is very low, fluids exsolved from the crystallizing magma (IF1 and W stage) were initially CO₂-rich followed by H₂O-rich fluids that carried chlorine and Au-associated metals (IF2 and Au/Bi/Te/As stage; Fig. 15D). This process ws already proposed by Baker and Lang (2001) on the . Emerald Lake and Dublin Gulch gold deposits (Yukon).

Outside E-W corridors and horsetail termination of “W” structures, magmatic fluids circulated along permeable horizons and reacted with specific reactive stratigraphic levels, mainly Viséan carbonates, forming skarns with disseminated gold and sulfides.

Results from radiogenic isotope analyses (Pb, Sr, Nd) indicate that the metals were derived from various sources that included the local granites and the surrounding Paleozoic rocks. The mobilization of lead from these rocks would have been facilitated by the large fracturing event related to the late Pb-Ag-(Zn) mineralization. Although the granites may have been a minor contributor to the formation of lead mineralization, they were likely a major source for W, Mo, Te, Au and Sn. Whether the basement contribution to the W-Au event was the result of remobilization by magma genesis or residual “crustal” metamorphic fluids remains uncertain. However, a “crustal” fluid flow might have been triggered by convection cells induced by the emplacement of the underlying pluton (Eldursi et al., 2009) (Fig. 15C).

The Pb-Ag-Zn veins reflect a convective system that was emplaced long after the Au-W mineralization (Jébrak, 1984; Wadjinny, 1998). The size of the convective system was large, extending over an area that was more than 10 times the size of the earlier three stages, and reflects a convective system large enough to homogenize lead isotope signatures at the district scale.

The superposition of the Pb-Zn-Ag hydrothermal system on the early W-Au system could be explained (1) by local uplift marked by their emplacement in a low hydrostatic pressure environment. The transtensional component of the D3 event is in need of more study, but major faults, such as the Tighza fault (Fig. 1, 3i), present numerous criteria of north-vergent normal sense of slip which might explain part of the observed uplift, or (2) by focusing of the fluid at the top of the solidified and cold intrusion, in relation with either a contrast of permeability or a contrast of thermal conductivity between the pluton and its host rocks. In the first hypothesis, gravity-induced circulation would move a single-pass fluid outward towards the periphery of the district, whereas multiple-pass fluids would circulate in-ward. The high salinity of the fluid and the lead isotopic signature are more compatible with the second model.

7.2 - Comparisons with the R-IRG model

This study of the Tighza district provides a better understanding of the conditions of emplacement and the dynamics of the formation of a late-Hercynian, polymetallic Au-As-W-Pb-Zn-Ba district. It has numerous characteristics that have been considered typical of R-IRG ore deposit model (Lang and Baker, 2001; Blevin, 2004; Hart, 2005).

7.2.1 - Comparison of the geological setting

The geology of IRG-type mineralization varies as a function of depth of emplacement resulting in a large diversity of geological environments including breccia, skarn, disseminated and vein-style deposits. They are associated with the presence of weakly reduced to moderately oxidised, intermediate to felsic fractionated I-type magmatism, either intermediate (i.e. granodiorite) or felsic. Granites generally fall into the field between ilmenite- and magnetite-series granites. They have subequal quartz, plagioclase and alkali feldspar with minor biotite, amphibole (biotite > amphibole), and apatite. Magnetite is rare or absent, due to low relative Fe content and/or weak oxidation state. High level rhyolites are strongly porphyritic. Granites are equigranular to porphyritic. Both are typically texturally variable and may present unidirectional solidification textures, miarolitic textures, pegmatites, vein dykes, all indicative of volatile saturation during crystallisation at mid to high crustal levels. Magmas are weakly reduced to moderately oxidised (FMQ to NNO). Alkali feldspars may be pinkish, but commonly white to greenish due to phyllic alteration proximal to mineralisation.

The Tighza district displays the same system of rapidly cooled polyphased intrusions as described in the IRG model. For example, the W-Mo event is directly associated with the Mine granite

(286.0 \pm 0.4 Ma) which is also the source of the W- skarns (285.3 \pm 0.5 Ma) and scheelite – molybdenite veins (285.6 \pm 0.5 Ma).

Both the host rocks and old structures within these rocks can constrain the deposition of the metals; reduced host rocks act as traps for oxidized hydrothermal fluids; existing structures may be reactivated during granite emplacement leading to the interplay of reactivated older structures and the evolving magmatic-hydrothermal system.

7.2.2 - Au Mineralization

Mineralisation is marked by an association of gold with Bi, As, W, Mo, Sb, Te. Base metal contents are highly variable; alteration is variable in both style and intensity. Zonation from the intrusion is from high temperature W-Mo (proximal) to later, lower temperature Au-As-Bi assemblages (distal). This latter assemblage occurs in vertically zoned porphyry systems and in longitudinal vein systems. Zonation on a pluton to district scale is also apparent. Gold mineralization in the Tighza district is restricted to the end of the magmatic stage. Moreover, its association with late sulfidic conditions may reflect either boiling of the hydrothermal fluids related to a change from lithostatic to hydrostatic pressure constraints, or an increase in sulfur released by the interaction of the host-rocks. Alteration styles comprise mostly a potassic (K-feldspar, muscovite and phlogopite) locally forming greisen with W-Au veins, and an intense silicification with skarn and skarnoid.

7.2.3 - Genesis

Fluid compositions are highly saline and of magmatic origin during early, pre-Au mineralisation stages. Gold mineralisation is typically associated with lower temperature, lower salinity fluids which are commonly CO₂ bearing, ranging from magmatic to dominantly magmatic with variable amounts of external fluids, probably introduced into the hydrothermal system during cooling.

Gold is typically associated with brines. Frank et al. (2002) demonstrated that gold is transported as AuCl_2H and $\text{Au}(\text{OH})$ in a sulfur-free magmatic volatile phase exsolved from haplogranite melts at 100 MPa and 800°C. Gold concentrations of magmatic vapor and brine converge as the critical pressure is approached along a given isotherm (Simon et al., 2005). The depth at which a melt undergoes volatile exsolution plays therefore a critical role in the mass transfer of gold between melt and magmatic volatile phases. Low S magmas of intermediate to felsic compositions with intermediate oxidation states favour neither early sulfide or magnetite precipitation, or early SO_2 formation. The absence of these conditions should be conducive to the preservation of Au in the melt fraction of granite magmas.

Table 5 – Comparison of major features between Tighza district and Hart's R-IRG model (2005)

Granite emplacement was preceded by reactivation of a shear-stress corridor that was invaded by fluids throughout the period of magmatic activity in the area, i.e. for at least 6 Ma (291.8 to 285 Ma). This suggests a context similar to that of Clear Creek, Yukon (Stephens et al., 2000). The long duration of cooling ($291.8 - 285.3 = 6.5$ Ma) implies conductive cooling of the pluton in a zone of low permeability (Cathles 1981).

8 - Conclusion

Mineralization within the Tighza polymetallic district was a multiphase process. Although the early gold-tungsten system and the large Pb-Zn-Ag vein-style system are both located at the top of a late-Hercynian plutonic system, they are clearly the product of two disconnected hydrothermal stages. This leads to propose some changes in the Lang and Baker model (2001) in its application

to the Tighza district, such as the coeval age of Au and Pb-Zn-Ag mineralization in most districts. In the Tighza district, the first stage was produced over a time span of at least 5 Ma by successive magmatic and hydrothermal pulses.

The geometry of the ore bodies is strongly dependent on the style of permeability induced in, and the reactivity of, the host rocks. For example, the Au-W quartz veins were deposited in veins formed by sinistral movement along a fault, while fluids reacting with limestone host-rock gave birth to the As-Fe-(Au) dissemination within skarns. Deposition of tungsten and molybdenum in veins and skarns is coeval with emplacement and cooling of the Mine granite at 286 ± 1 Ma.

This polyphased emplacement was accompanied by uplift of the basement. This conclusion is supported by field observation of deformed host-rocks, re-equilibration and spectacular explosion of early water-carbonic inclusions of the tungsten phase. This uplift is coeval with the appearance of saccharoidal quartz with multiphase fluid inclusions oversaturated in NaCl. This second stage is a late-magmatic mineralized episode that precedes the emplacement of large hydrothermal convective cells of regional extent, responsible for the genesis of huge Pb-Ag-(Zn) veins, likely during the upper Permian.

The late-Hercynian gold veins of central Morocco differ from gold-veins associated with the Hercynian terranes of France in multiple aspects: structural context, fluid origins, isotopic signatures, and younger ages (Bouchot et al., 1997, 2000, 2005). Centered on a calc-alkaline reduced intrusion, the Tighza district thus appears to be the missing link which allows an association to be made between three apparently disconnected types of ore deposits within a single district: 1) W-greissens, somewhat acid and less –developed; 2) reduced gold deposits; and 3) lead-zinc-silver deposits in sediments (Beaudoin and Sangster, 1992).

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Annexe 1: Methodologies

Fluid inclusions

Methodology

Fluid inclusions were carried out using a Chaix-Meca system. Low temperature standards included Campeirio quartz, sea-water and pure synthetic fluid inclusions whereas high temperature standards consisted of lead orthophosphate and a critical water biphasic fluid inclusion in a synthetic quartz. After calibration, precision is $<0.1^{\circ}\text{C}$ (-60° – 0°C range) for low temperatures whereas it is $<1^{\circ}\text{C}$ for high temperatures (200° – 400°C range). Cycled heating and cooling were used for observing clathrate formation and dissolution. Raman spectroscopy was used for the determination of solids, fluids and gas within fluid inclusions.

Radiogenic isotopes

Rb-Sr, Sm-Nd

Rb-Sr, Sm-Nd isotopic data were obtained from scheelite and whole-rock samples analyzed at the GEOTOP Radiogenic Isotope Laboratories at the Université du Québec à Montréal (UQAM) in order to constrain the sources of the mineralization (Voicu et al., 2000, Kempe et al., 2001). Scheelite was concentrated from vein material by passing crushed vein material through heavy liquids, magnetic separation and, finally hand-picking under a binocular microscope. The resulting mineral fraction was washed in dilute hydrochloric acid (HCl) to remove any carbonate material, then rinsed in distilled water to remove the acid and left to dry. The scheelite fractions were ground to a fine powder in an agate mortar and whole-rock samples were similarly ground in an agate puck-mill. Chemical separation procedures for Nd and Sr isotopes were conducted under clean lab conditions and all acids were distilled. In each case, between 50 and 100 mg of powdered

sample were weighed out into a Teflon Parr-bomb and spiked with a $^{150}\text{Nd} - ^{149}\text{Sm}$ tracer solution for the determination of Sm and Nd concentrations. A mixture of HF-HNO₃ acids was added and the mixture placed in an oven in order to dissolve the samples under pressure at a temperature of 150°C. After one week, the samples were removed from the furnace and evaporated. The samples were subsequently evaporated in perchloric acid to break up the fluoride salts and redissolved in 6 M HCL in the oven again for 12 hours. The resulting 6 M HCL solution was loaded onto ion-exchange columns containing AG1X8 resin that retained the Fe in the sample but allowing the other elements to be eluted with 6 M HCL. The samples were evaporated and 0.5 ml of 14M HNO₃ was added to change the HCl salts into nitrates. The REE and Sr separation requires an Fe-free solution. This was achieved using by passing the 6 N HCl solutions through an anion exchange resin (AG X-1) that retains the iron. Sr and REE fractions were then concentrated using Eichrom TRU Spec resin. Sr was subsequently purified using Eichrom Sr Spec resin and Sm and Nd separation was achieved using columns containing Eichrom LN Spec resin.

The isotopic composition and concentration of Nd, Sm, and Sr were analysed by thermal ionisation mass spectrometry (TIMS). Nd and Sm were measured using a triple filament assemblage with the samples evaporated on Ta side filaments and ionized by a Re centre filament. Sr samples were loaded and analysed on a single Re centre filament. The Nd and Sr samples were measured in dynamic mode and normalized to $^{146}\text{Nd}/^{144}\text{Nd} = 0.7219$ and $^{86}\text{Sr}/^{88}\text{Sr} = 0.1194$ assuming exponential fractionation behaviour while Sm was measured in static mode. During the period of this study, repeated measurements of the Nd International Standard JNdi yielded a value of $^{143}\text{Nd}/^{144}\text{Nd} = 0.512148 \pm 11$ (N=11) compared to the published value of 0.512115 ± 7 (Tanaka et al., 2000). Our samples were corrected by 0.000033 for taking into account the shift of the measured standard. Repeated analysis of the NBS 987 Sr standard yielded a value of $^{87}\text{Sr}/^{88}\text{Sr} = 0.710263 \pm 17$ (N=7). Epsilon Nd values were calculated using the present day CHUR value of

0.512635 for $^{143}\text{Nd}/^{144}\text{Nd}$ and a $^{147}\text{Sm}/^{144}\text{Nd} = 0.1967$. The typical combined procedure blanks for Nd and Sm are <150 pg.

Ar/Ar

Four samples of phyllosilicates, two muscovites from a scheelite skarn (NK 222) and a molybdenite-scheelite vein (NK 220), both from underground works, and two biotites from Mine granite (NK 211) and W1 north vein (NK 313) were analysed with a ^{39}Ar - ^{40}Ar laser probe (CO_2 Synrad®). Two other samples (muscovite from W1 north vein and from greisen intragranite) gave no exploitable results.

Individual mineral grains were handpicked under a binocular microscope from crushed rocks (0.3-2 mm fraction). Selected grains were wrapped in Al foil to form packets (11 mm \times 11 mm \times 0.5 mm). These packets were stacked up to form a pile, within which packets of flux monitors were inserted every 10 samples. The stack was inserted in an irradiation can and was irradiated at the McMaster reactor (Hamilton, Canada). The irradiation standard was sanidine TCR-2 (28.34 Ma according to Renne et al., 1998). The sample arrangement allowed us to monitor the flux gradient with a precision of $\pm 0.2\%$.

The step-heating experimental procedure is described in detail in Ruffet et al. (1995, 1997). Blanks are performed routinely each first or third run, and are subtracted from the subsequent sample gas fractions. Analyses are performed on a Map215® mass spectrometer. To define a plateau age, a minimum of three consecutive steps are required, corresponding to a minimum of 70% of the total ^{39}ArK released, and the individual fraction ages should agree to within 1σ or 2σ with the integrated age of the plateau segment. All discussed ^{39}Ar - ^{40}Ar results are displayed at the 1σ level. Pseudo-plateau ages, with less than 70% of ^{39}ArK released, can also be defined (Table 2).

Pb

Analytical procedures are summarized herein. Between 10 and 20 milligrams of sulfide powder was dissolved in 7N HNO₃. The residue was taken up in 1N HBr for Pb chemistry. Pb was separated in Bio-Rad 10-ml polyethylene columns and Dowex AG1-8X anion resin, using 1N HBR to elute other elements and 6N HCl to elute Pb. Total procedural blanks for Pb are <250 picograms. Samples were loaded onto single Re filaments with H₃PO₄ and silica gel, and were run at filament temperatures of 1175-1225°C. Analysis of USGS Standard BCR-1 yields Pb = 13.56 ppm, U = 1.70 ppm, Th = 5.86 ppm, ²⁰⁶Pb/²⁰⁴Pb = 18.818, ²⁰⁷Pb/²⁰⁴Pb = 15.633, and ²⁰⁸Pb/²⁰⁴Pb = 38.633 (average of 2 runs, see Tatsumoto et al., 1972). All mass spectrometer runs were corrected for fractionation using NIST SRM981. The average ratios measured for SRM981 on the new Triton Mass spectrometer are ²⁰⁶Pb/²⁰⁴Pb = 16.892 ± 0.010, ²⁰⁷Pb/²⁰⁴Pb = 15.432 ± 0.013, and ²⁰⁸Pb/²⁰⁴Pb = 36.512 ± 0.038 (2 s.d.), based on 15 runs. The fractionation correction, based on the values of Todt et al. (1984) is +0.13% amu.

Annexe 2 – Microprobe analysis of arsenopyrite, pyrrhotite, loellingite and sphalerite from Tighza. Microprobe CAMEBAX SX 50, ISTO – BRGM laboratory, Orléans, France

	arsenopyrite									sphalerite			
	pyrrhotite vein (mine) (19)	FNJ4 and 4 bis ddh (16)	As-Fe-(Au) dissemination s (SIH-4 ddh) (9)	As-Fe-(Au) dissemination s - veinlets (SIH-4 ddh) (5)	pyrrhotite vein (skarn of Mispickel granite) (15)	W1 vein (surface) (15)	W1 north vein (surface) (10)	W1 vein (drill core) (7)	W4 vein (SW4J-4 ddh) (6)		pyrrhotite vein (mine, level 857) (13)	pyrrhotite vein (mine, level 707) (18)	W4 vein (ddh SW4J-4) (12)
S	18.20	19.83	19.30	20.668	18.78	19.62	18.17	18.36	18.90	S	33.62	33.76	33.43
Fe	33.50	33.70	33.56	34.398	34.26	33.38	32.56	34.64	34.80	Fe	9.27	10.72	11.37
As	47.56	45.27	46.48	44.94	46.89	45.29	47.39	48.64	47.90	Bi	0.08	0.07	0.08
Co	0.16	0.54	0.01	0.00	0.02	0.25	1.01	0.09	0.14	Cu	0.05	0.01	0.00
Ni	0.01	0.02	0.01	0.00	0.00	0.41	0.01	0.00	0.00	Cd	0.14	0.37	0.59
	99.48	99.42	99.57	100.04	100.00	99.04	99.27	101.73	101.72	Zn	56.14	54.39	55.20
											99.30	99.33	100.66
S cat	31.43	33.68	32.97	34.64	32.07	33.50	31.47	31.03	31.76				
As cat	35.15	32.91	33.99	32.24	34.28	33.11	35.14	35.22	34.46	môle % FeS	14.86	17.23	17.86
Fe cat	33.22	32.86	32.92	33.10	33.60	32.73	32.38	33.63	33.58				
	loellingite					pyrrhotite							
	W1 north vein (surface) (11)	W1 north vein (drill core) (6)	W4 vein (drill core) (5)	As-Fe-(Au) dissemination (skarn of	pyrrhotite vein (skarn of Mispickel	pyrrhotite vein (mine) (8)		pyrrhotite vein	As-Fe-(Au) dissemination				W-Au quartz veins

				Kaolin granit e) (4)	kel granit e) (3)								
								mine (29)	skarn of Kaolin granit e (ddh S II-4) (12)	skarn of Kaolin granit e (surfac e) (4)	Mispic kel granit e (ddh FNJ-4) (13)	Mispic kel granit e, surfac e (9)	surfac e (13)
S	1.65	1.72	2.01	0.55	1.98	2.34							
Fe	26.79	28.57	28.57	14.80	28.64	27.52	S anion	54.16	54.28	54.16	54.15	53.23	53.36
As	69.64	70.81	71.09	70.12	69.28	69.80	(Fe+Co +Ni) cation	45.77	45.62	45.77	45.78	46.71	46.56
Co	0.96	0.02	0.31	4.05	0.00	0.35	formul a (S = 8)	Fe _{6.76} S ₈	Fe _{6.72} S ₈	Fe _{6.76} S ₈	Fe _{6.76} S ₈	Fe _{7.02} S ₈	Fe _{6.98} S ₈
Ni	0.00	0.00	0.13	9.57	0.00	0.30							
	99.05	101.12	102.11	99.09	99.90	100.31							
	Bismuth sulfides (W1 north vein, ddh SW1N 3b, 56m)												
	joséite B	joséite B	joséite B	joséite B	joséite B	hedleyi te	hedleyi te	hedleyi te					
	2.92	2.90	2.59	2.81	2.81	0.00	0.07	0.02					
S	0.24	0.10	0.09	0.24	0.07	0.22	0.28	0.22					
Sb	0.04	0.00	0.03	0.27	0.10	0.12	0.05	0.13					
Se	75.33	74.06	76.62	74.89	72.99	80.42	77.51	79.87					
Bi	22.23	21.90	19.95	21.60	22.35	19.70	22.19	18.62					
Te	100.76	98.96	99.28	99.81	98.32	100.46	100.10	98.86					

Sample	Analysed mineral	% AP	Conventionnel age
Molybdenite vein (mine level 707 (NK 220))	Muscovite	98,3	285.6 ± 0.5 Ma
W-skarnoid (mine level 707) (NK 222)	Muscovite	97,3	285.3 ± 0.5 Ma
Mine granite (surface) (NK 211)	Biotite	98,9	286.0 ± 0.4 Ma
W1 north vein (surface gallery) (NK 313)	Biotite	91.4	291.8 ± 0.3 Ma

Sample	Analyzed material	$^{206}\text{Pb} / ^{204}\text{Pb}$	$^{207}\text{Pb} / ^{204}\text{Pb}$	$^{208}\text{Pb} / ^{204}\text{Pb}$
Au-W veins				
W1 vein	Pb-rich polymetallic sulphides	18.298	15.668	38.497
W1 North vein	Pb-rich polymetallic sulphides	18.351	15.669	38.383
W1 North vein	Pb-rich polymetallic sulphides	18.343	15.660	38.425
W1 North vein	Pb-rich polymetallic sulphides	18.335	15.629	38.329
W5 vein	Pb-rich polymetallic sulphides	18.286	15.680	38.534
W5 vein	Pb-rich polymetallic sulphides	18.417	15.653	38.428

W5 vein (duplicata)	Pb-rich polymetallic sulphides	18.419	15.626	38.359
W1 vein	Pb-rich polymetallic sulphides	18.371	15.640	38.329
Pyrrhotite vein				
mine level 707	Pb-rich polymetallic sulphides	18.261	15.646	38.427
mine level 707	Pb-rich polymetallic sulphides	18.273	15.661	38.476
Mispickel granite	Pb-rich polymetallic sulphides	18.266	15.658	38.423
W-skarnoid				
scheelite ore (mine level 757)	Pb-rich polymetallic sulphides	18.694	15.671	38.921

scheelite ore (mine level 757)	Pb-rich polymetallic sulphides	18.875	15.686	38.613
As-Fe-(Au) dissemination				
Kaolin granite (drill-hole S-II-4 136 m)	Pb-rich polymetallic sulphides	18.299	15.643	38.490
Kaolin granite (drill-hole S-II-4 133 m)	Pb-rich polymetallic sulphides	18.323	15.636	38.568
Kaolin granite (drill hole S-II-4 138 m)	Pb-rich polymetallic sulphides	18.297	15.642	38.504
Mispickel granite (drill- hole FNJ-4bis 95 m)	Pb-rich polymetallic sulphides	18.319	15.667	38.582
Mispickel granite (drill-hole FNJ-4 298 m)	Pb-rich polymetallic sulphides	18.343	15.656	38.608
Pb-Ag-Zn veins				

Sidi Ahmed vein	galena	18.291	15.689	38.559
Sidi Ahmed vein	galena	18.264	15.676	38.519
Sidi Ahmed vein	galena	18.294	15.710	38.623
barytine vein	galena	18.274	15.673	38.518
Signal vein (1)	galena	18.261	15.682	38.508
Signal vein (1)	galena	18.259	15.675	38.485
Signal vein (1)	galena	18.248	15.661	38.439
Signal vein (1)	galena	18.283	15.696	38.556
Sidi Ahmed vein (1)	galena	18.232	15.652	38.412
Iguer Oujna vein (1)	galena	18.257	15.651	38.404
Rocks				
Mine granite	K-feldspar	18.300	15.704	38.621
Mispickel granite	K-feldspar	18.280	15.696	38.583

sample	identification	Nd	Sm	$^{147}\text{Nd}/^{144}\text{Nd}$	$^{143}\text{Nd}/^{144}\text{Nd}$	error	$\epsilon\text{Nd}(0)$	$\epsilon\text{Nd}(t)$	Tdm	$^{87}\text{Sr}/^{86}\text{Sr}$	error
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whole rock

NK216a (b7)	granite mispickel	28.3	5.04	0.1078	0.512333	0.000012	-6.0	-2.5	1.2	0.70420 ^a	0.00002
NK211 (b6)	Mine granite	21.9	4.01	0.1109	0.512451	0.000016	-3.6	-0.4	1.0	0.70997 ^b	0.00001
NK210 (b4)	greisen	25.0	3.98	0.0963	0.512402	0.000008	-4.6	-0.8	1.0		
NK206 (b3)	microtrondhjemite	25.5	4.52	0.1069	0.512416	0.000010	-4.3	-0.9	1.0	0.70802 ^c	0.00002
NK204 (b2)	rhyolite	31.0	6.25	0.1217	0.512302	0.000011	-6.6	-3.7	1.4		

scheelite

NK202 (b31)	W5 Au-W vein	311	103	0.1995	0.512277	0.000007	-7.0	-7.2		0.71364	0.00002
NK203 (b32)	W4 Au-W vein	380	129	0.2056	0.512220	0.000008	-8.2	-8.5		0.71038	0.00001
NK208 (b33)	WO3 Au-W vein	215	77.7	0.2183	0.512289	0.000010	-6.8	-7.6		0.71219	0.00002
NK218 (b34)#	stockwerk W	334	119	0.2145	0.512329	0.000010	-6.0	-6.7		0.71118	0.00002
NK220 (b35)*	W-Mo quartz vein	636	196	0.1862	0.512513	0.000009	-2.4	-2.0		0.70897	0.00002
NK221 (b36)*	W-skarn vein	577	198	0.2071	0.512297	0.000009	-6.7	-7.1		0.71122	0.00002
NK222 (b37)*	W-skarn vein	654	213	0.1971	0.512246	0.000007	-7.6	-7.7		0.71194	0.00002

a: 297 ppm Sr; 382 ppm Rb; 3.715: $^{85}\text{Rb}/^{86}\text{Sr}$; 0.72006: $^{87}\text{Sr}/^{86}\text{Sr}$ measured

b: 266 ppm Sr; 124 ppm Rb; 1.356: $^{85}\text{Rb}/^{86}\text{Sr}$; 0.71576: $^{87}\text{Sr}/^{86}\text{Sr}$ measured

c: 301 ppm Sr; 245 ppm Rb; 2.361: $^{85}\text{Rb}/^{86}\text{Sr}$; 0.71809: $^{87}\text{Sr}/^{86}\text{Sr}$ measured

*Mine level 757; # mispickel granite
age for calculation is in billion years

Features	Common to R-IRGD (compiled from Hart, 2005; Lang and Baker, 2001; Thompson et al., 1999)	Tighza (this study)	
		Y/N	specific characteristics and comments
Tectonic setting	<p>Ancient continental margins behind accretionary or collisional orogens and subduction-related magmatic arcs</p> <p>Preferred host strata include reducing basinal miogeoclinal sedimentary or metasedimentary rocks</p> <p>Intrusions are undeformed as they were intruded millions of years after regional deformation</p>	√	Location within the Mesetan variscan bloc which corresponds to an external zone (characterized by HT-LP metamorphism) of the variscan orogeny (Michard et al., 2008)
Intrusion	<p>Metaluminous, subalkalic intrusion of intermediate to felsic compositions that lie near the boundary between ilmenite and magnetite series</p> <p>Mineralizing plutons have “smoking gun” characteristics that indicate the likelihood of generation of hydrothermal fluids. Physical features and geochemical support should exist for high volatile contents, fluid exsolution, evidence of rapid fractionation, zoned plutons, porphyry textures, presence of aplite and pegmatite dykes, quartz and tourmaline veins, greisen alteration, miarolitic cavities and/or unidirectional-solidification textures, preferably in the plutons’ apexes</p>	√	Many similarities, among which calc-alkaline granite belong to ilmenite-series, and greisen alteration

	Reduced IRGS are associated with felsic, ilmenite-series plutons that lack magnetite, have low magnetic susceptibilities and aeromagnetic response, and have low ferric:ferrous ratios of <0.3. These types of plutons are uncommon in arc and fore-arc settings where orogenic gold deposits are most common		
Mineralization	<p>Metal assemblage that variably combines gold with elevated Bi, W, As, Mo, Te, and/or Sb and low concentrations of base metals</p> <p>Low sulphide mineral content, mostly <5 vol%, with a reduced ore mineral assemblage that typically comprises arsenopyrite, pyrrhotite and pyrite and lacks iron oxides</p>	√	<p>Similar metal assemblage with Bi, W, As, Mo, Te and few base metals</p> <p>Sulfide mineral content < 5 %</p>
Gold grade	Generally low, from 0.8 to 1.5 gr/t	√	Medium Au grade measured in the inner W and Au circle around the intrusion is about 0.7 g/t (fig. 4)
Associated mineralizations and metal zoning	Location in magmatic provinces best or formerly known for tungsten and/or tin deposits	√	Tin and tungsten granites formerly mined in Central

			Morocco
	Thermal gradients surrounding cooling plutons are steep and result in temperature-dependent concentric metal zones that develop outward from pluton margins for distances up to a few kilometres, or just beyond the thermal aureole. Pluton-proximal gold mineralization may be associated with Bi, Te, and W aureole-hosted mineralization will have an As or Sb tenor, and distal mineralization may be related to Ag-Pb-Zn	√	Same thermal and zonation patterns (fig. 2,3)
Fluids	Carbonic hydrothermal fluids	√	idem
Hydrothermal alteration	Weak	√	Weak and of few extension
Deposits type and structures	Wide range of mineralization styles around and/or within the intrusion: skarns, replacements, disseminations, stockworks and veins	√	Wide range of mineralization
	Sheeted Veins The most distinctive style of gold mineralization are sheeted arrays of parallel, low-sulphide, single-stage quartz veins which are found over 10s to 100s of metres and preferentially located in the pluton's cupola (Fig. 3). These veins are unlike multidirectional interconnected stockworks characteristic of porphyry systems or antithetic tensional vein arrays typical of orogenic gold deposits	√	Wide Au-quartz veins instead of sheeted Au-veins due to the pre-existing presence of faults
Timing	Intrusion-related deposits are coeval (± 2 m.y.) with their associated, causative pluton.	√	Same age for both intrusion-related deposit and

			associated pluton (286 ± 1 Ma)
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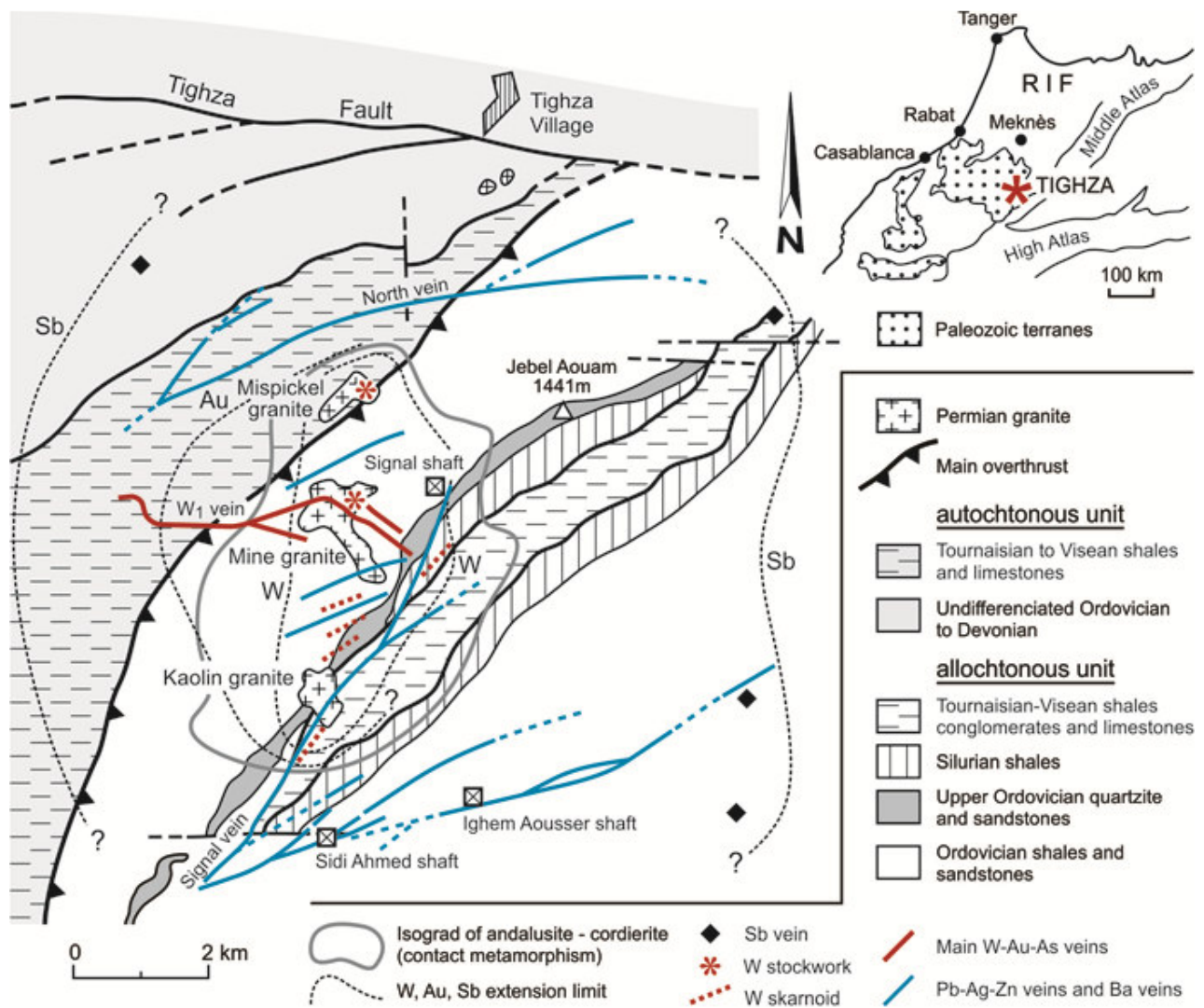


figure n°1

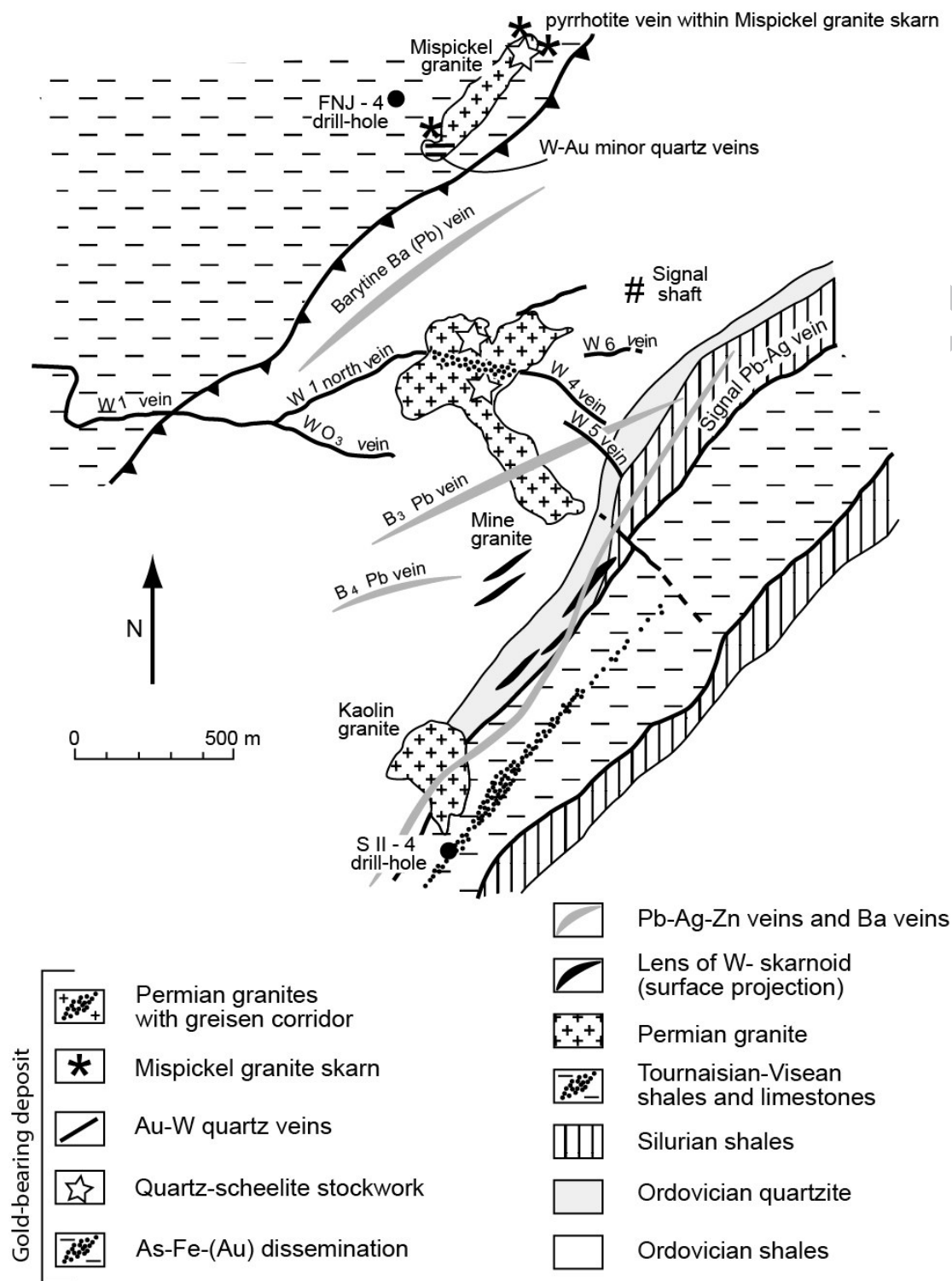
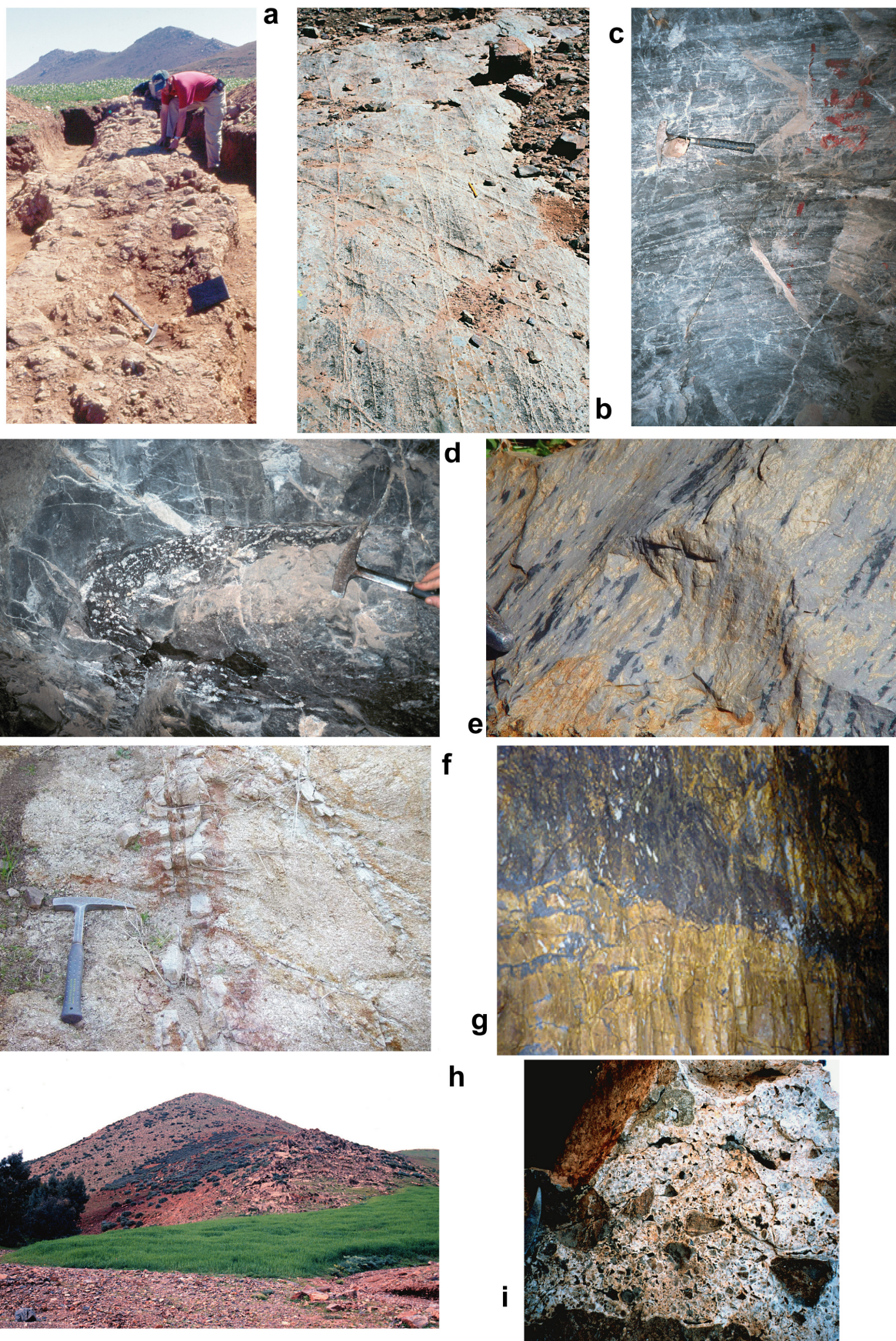
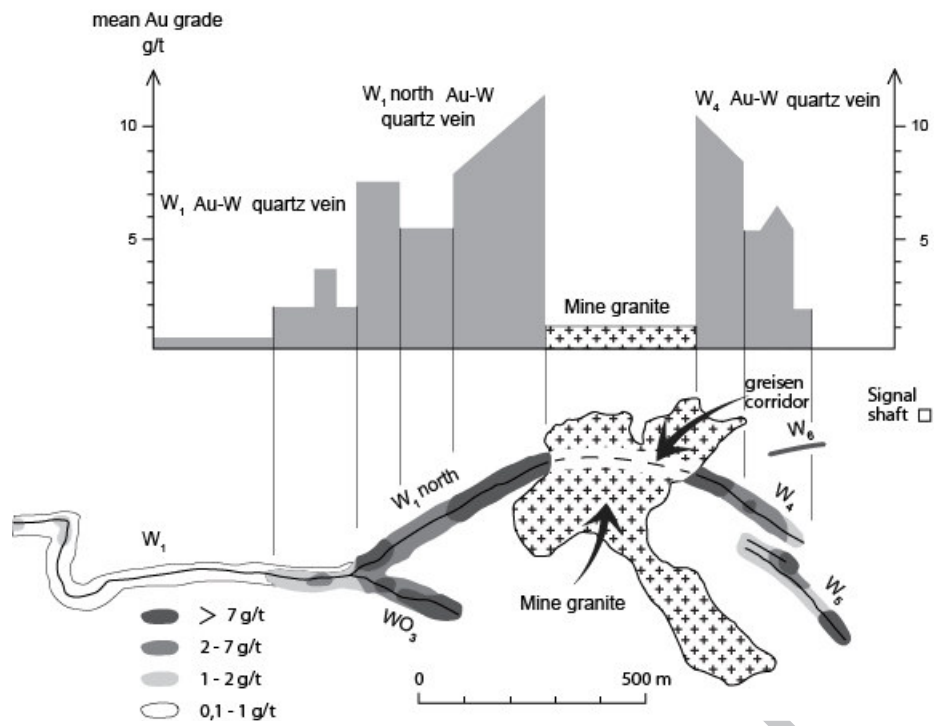
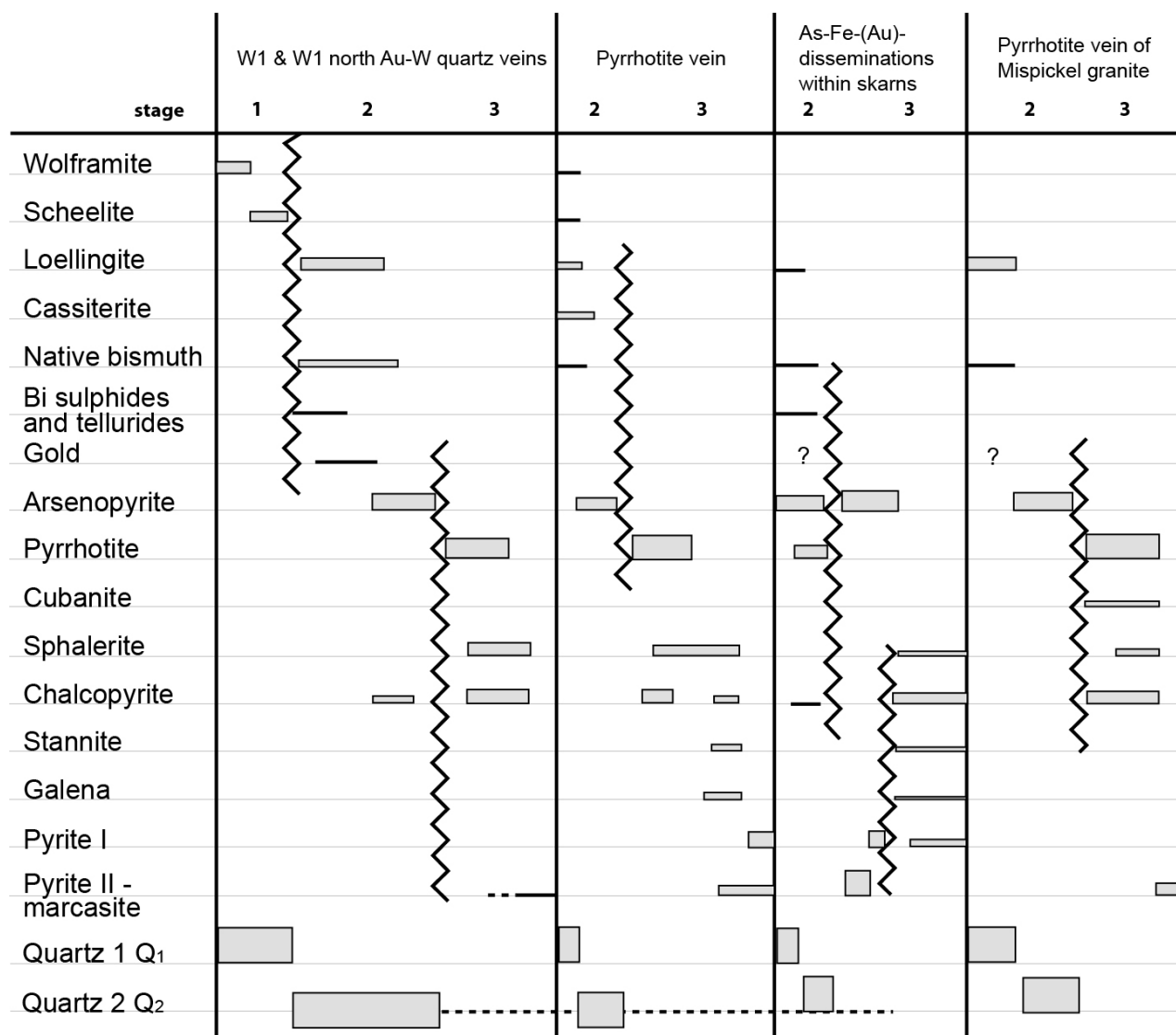
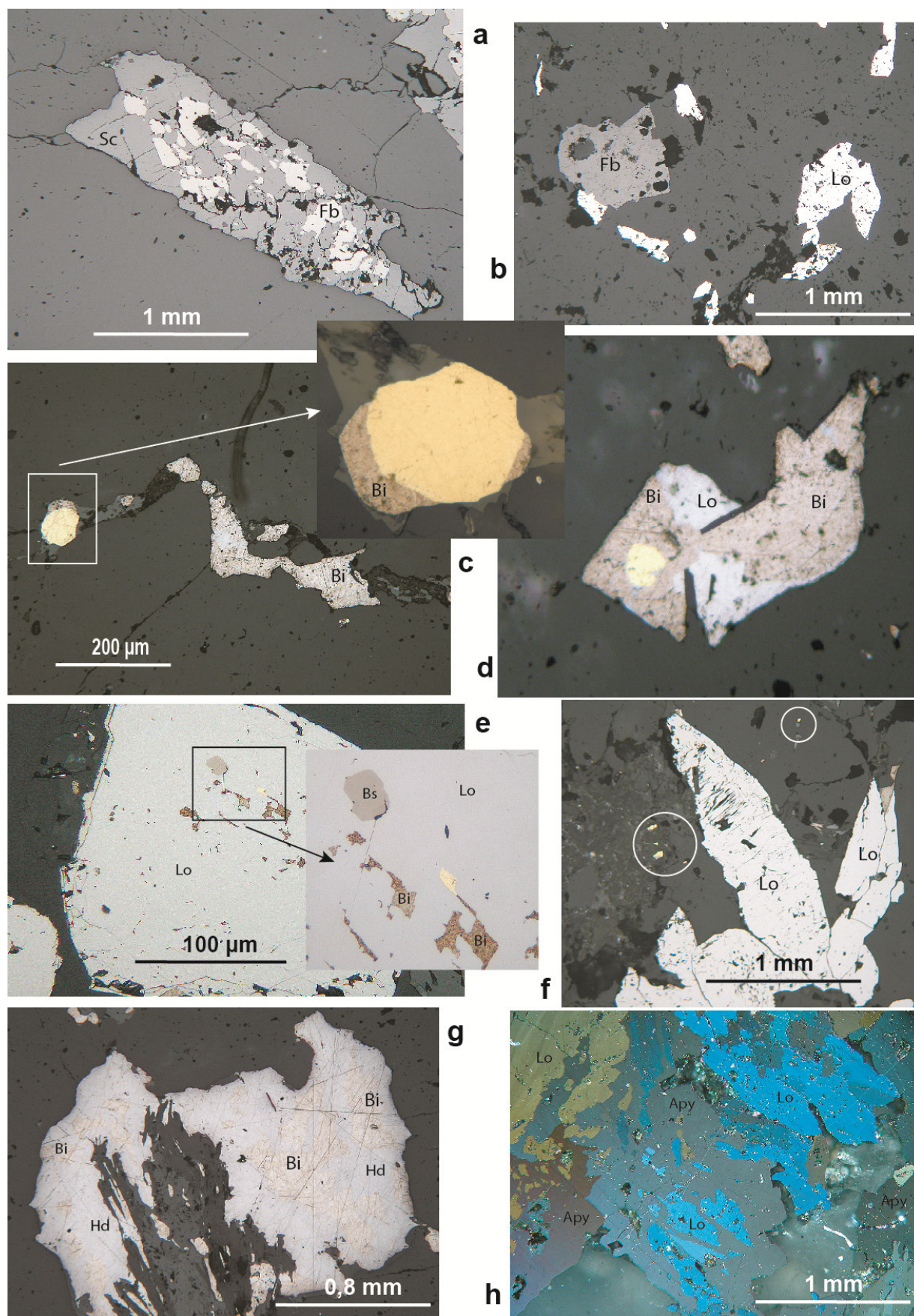


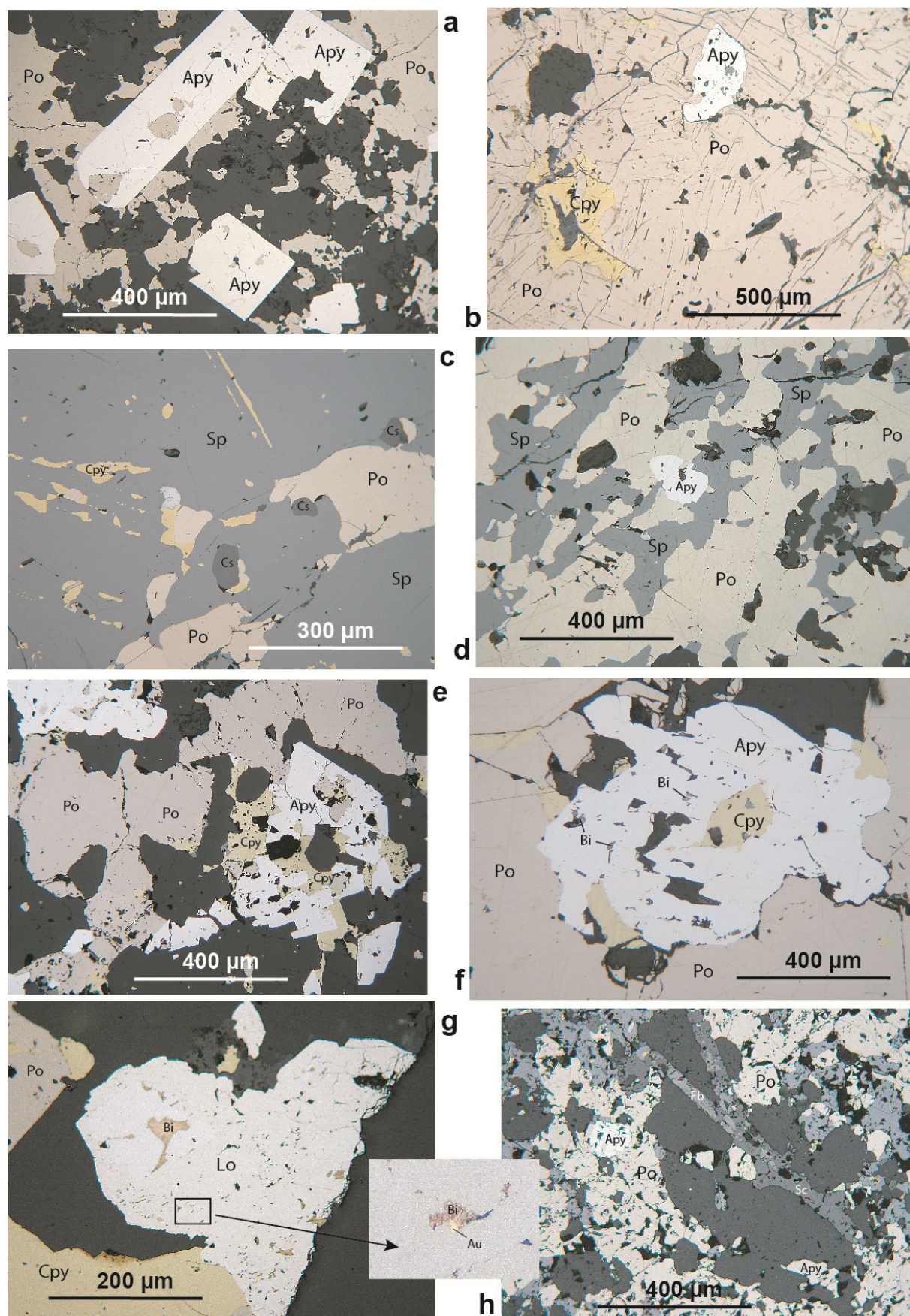
figure n°2



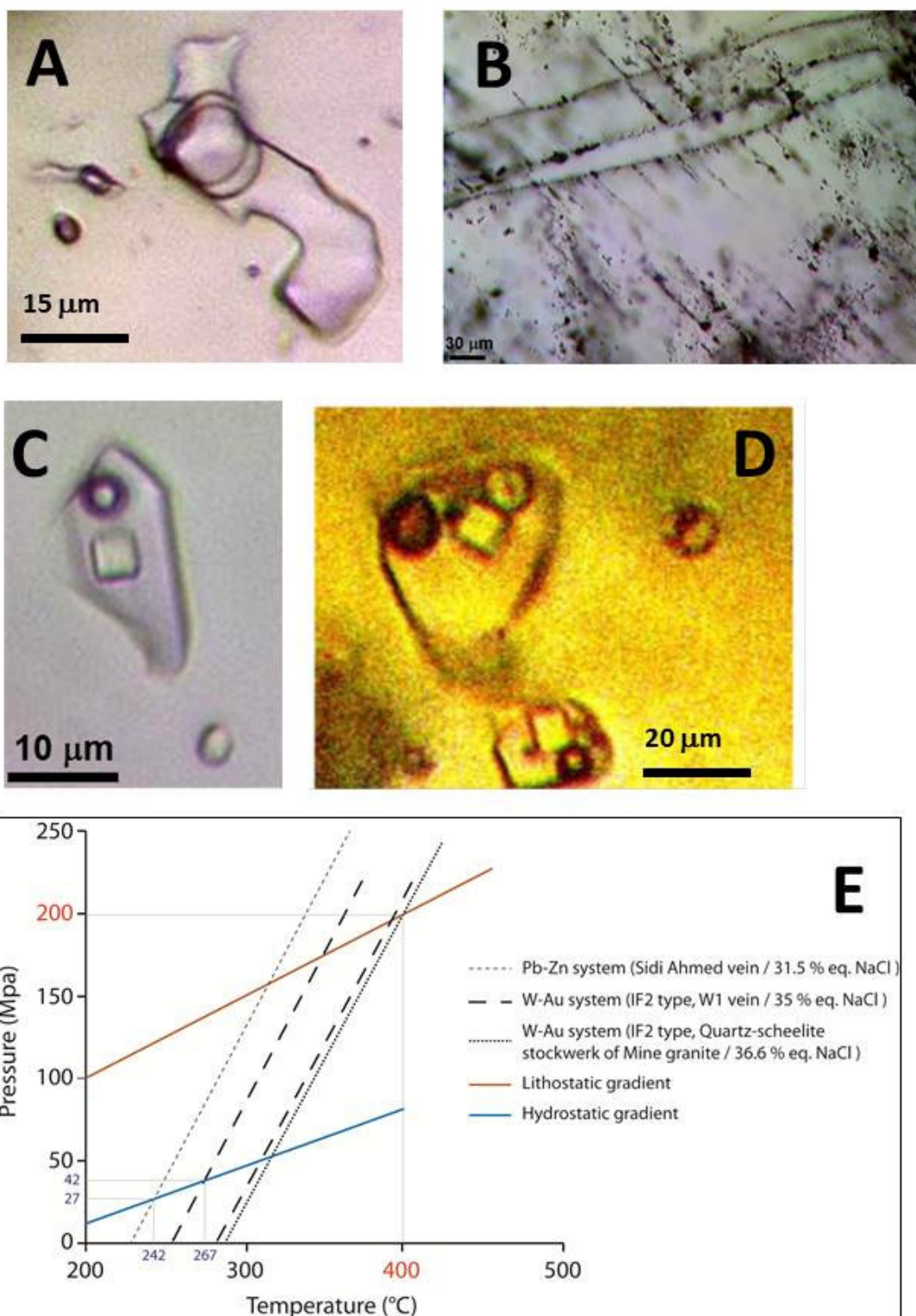


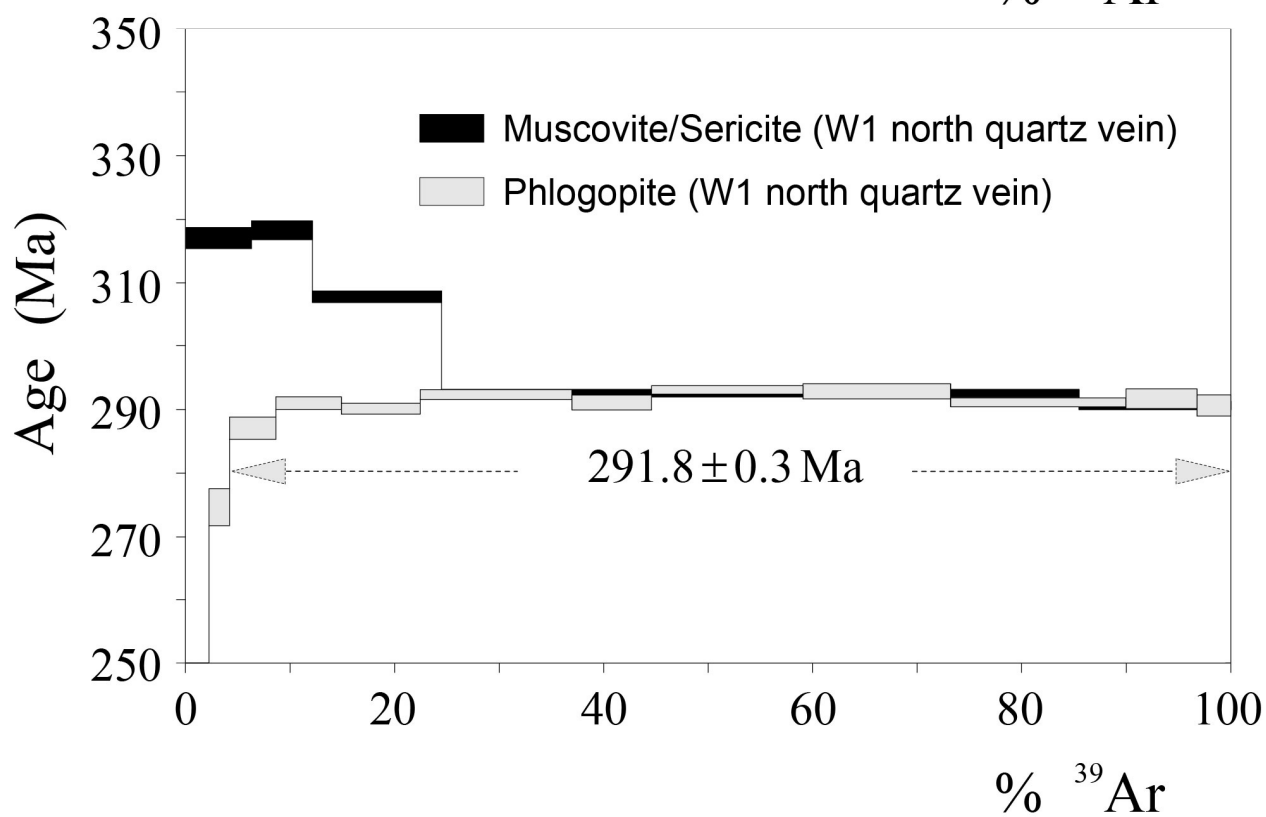
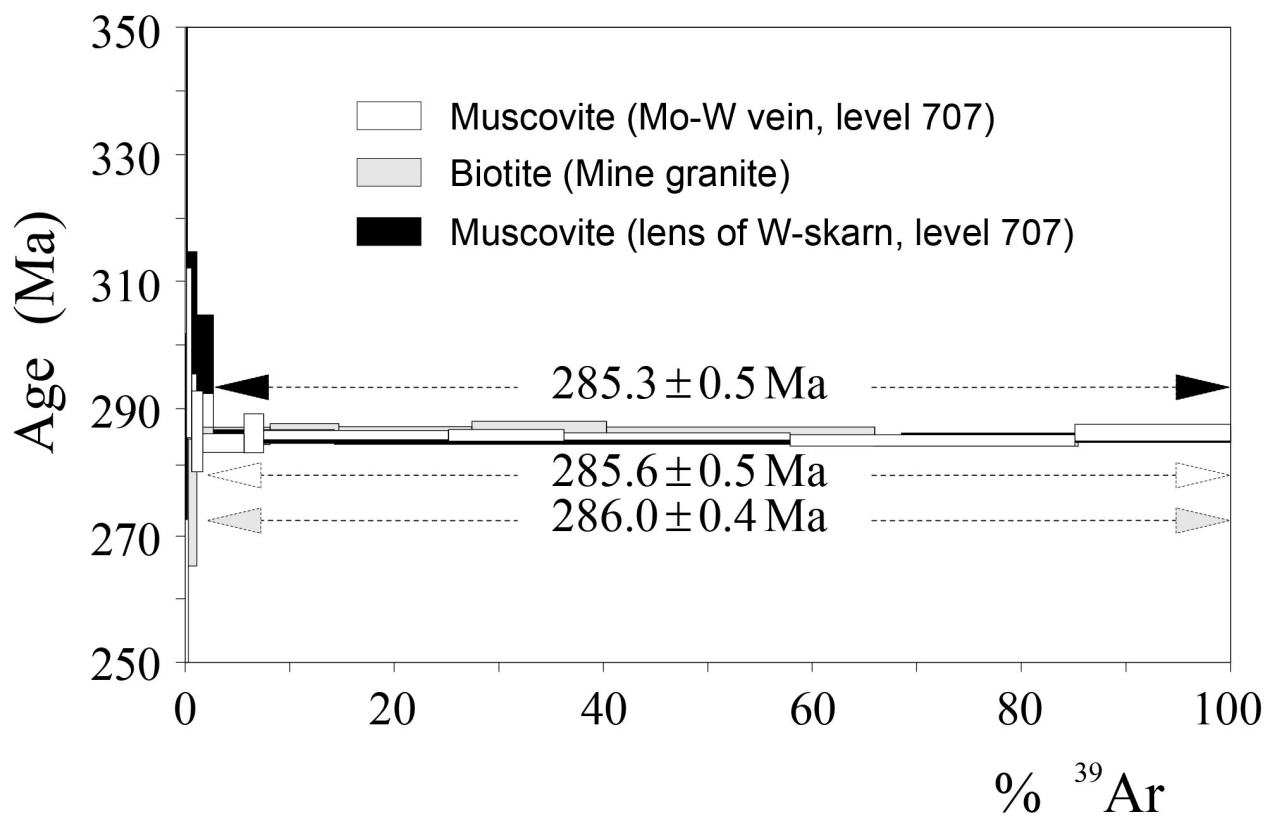


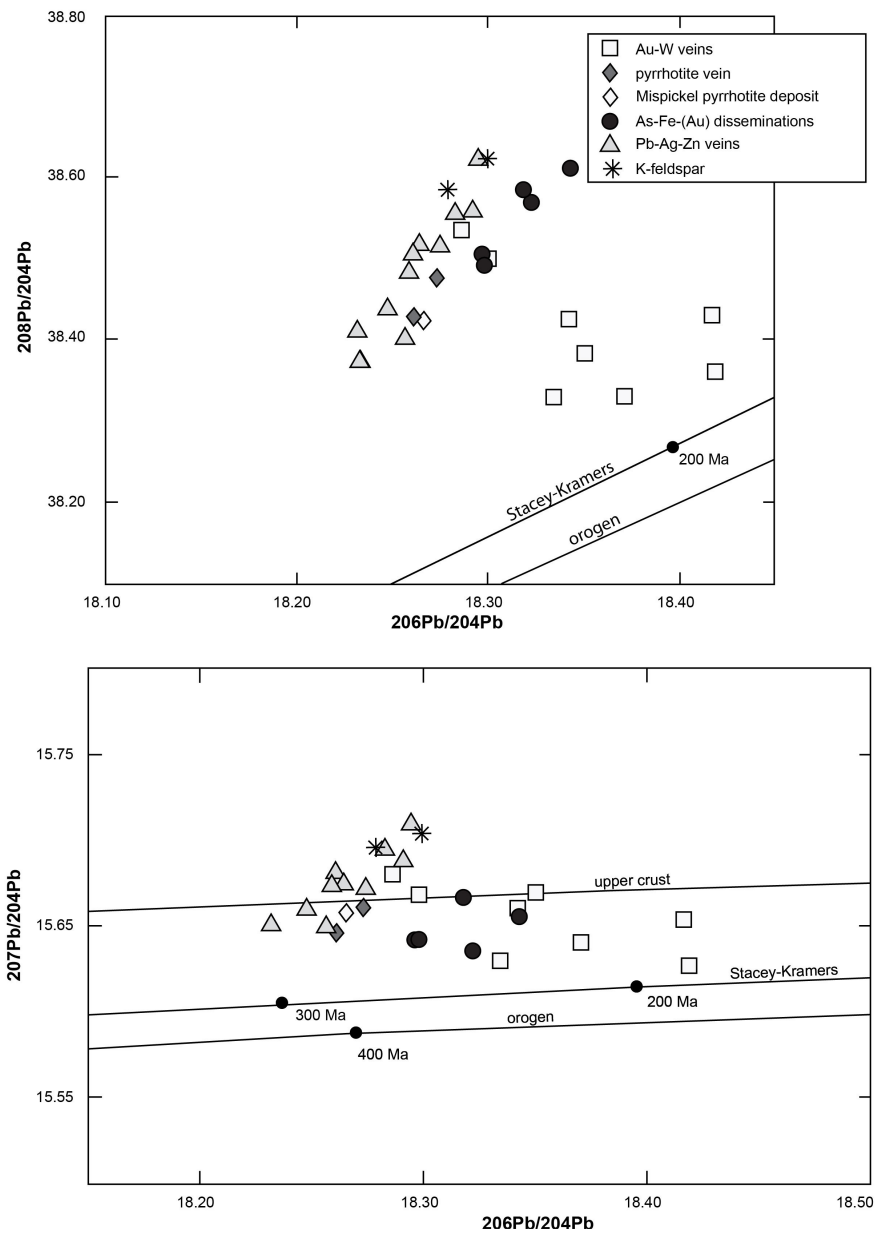


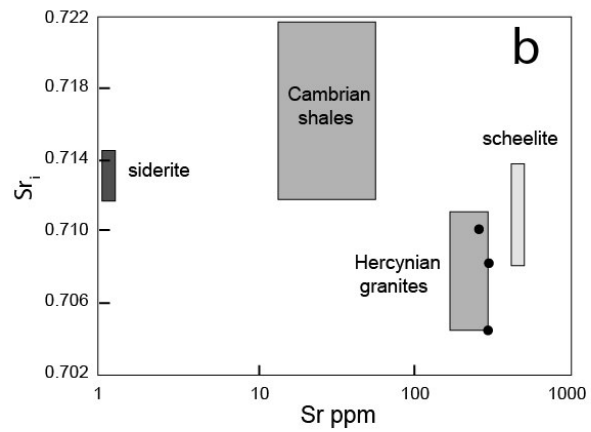
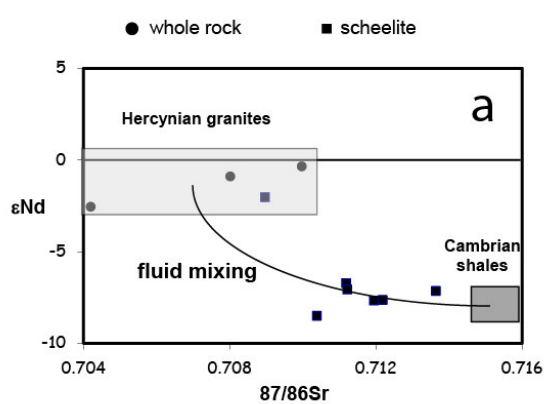


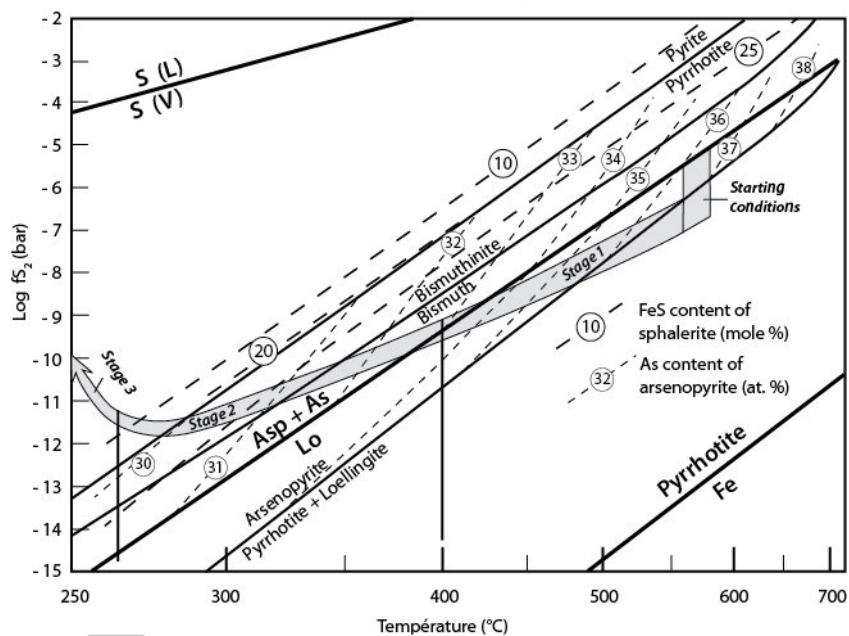
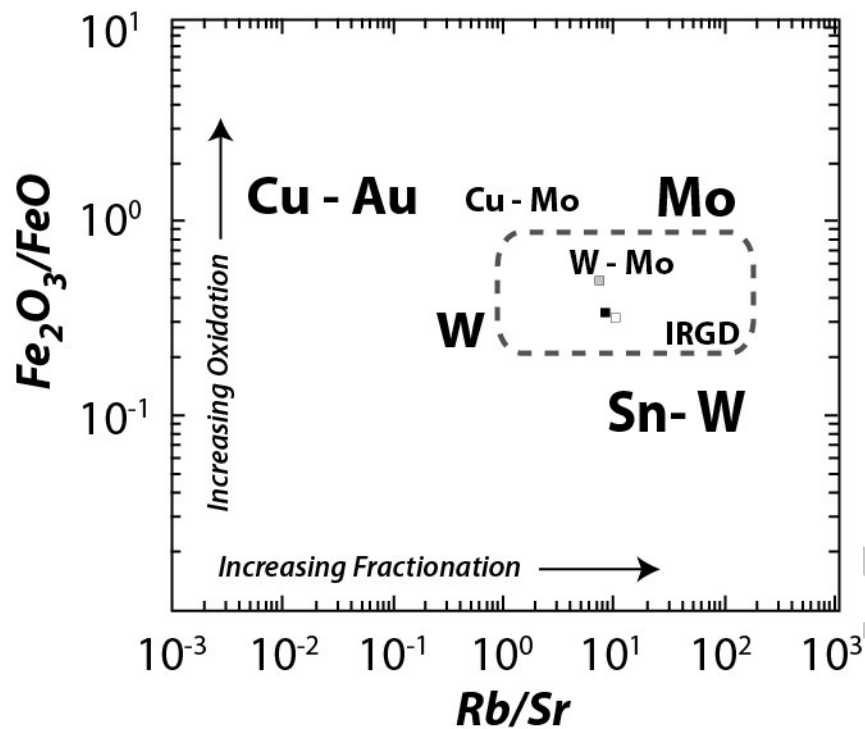


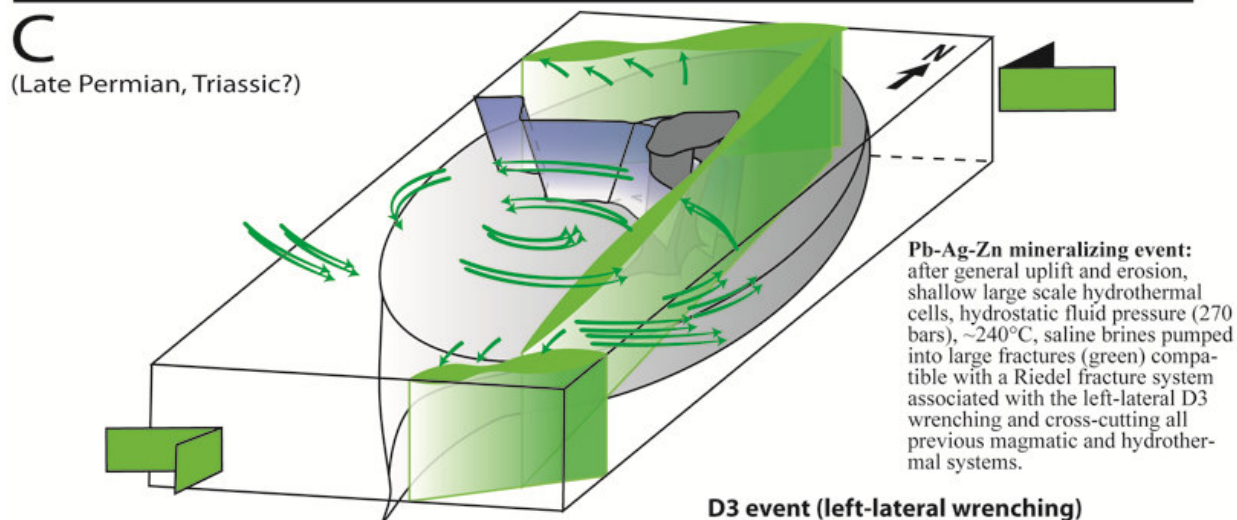
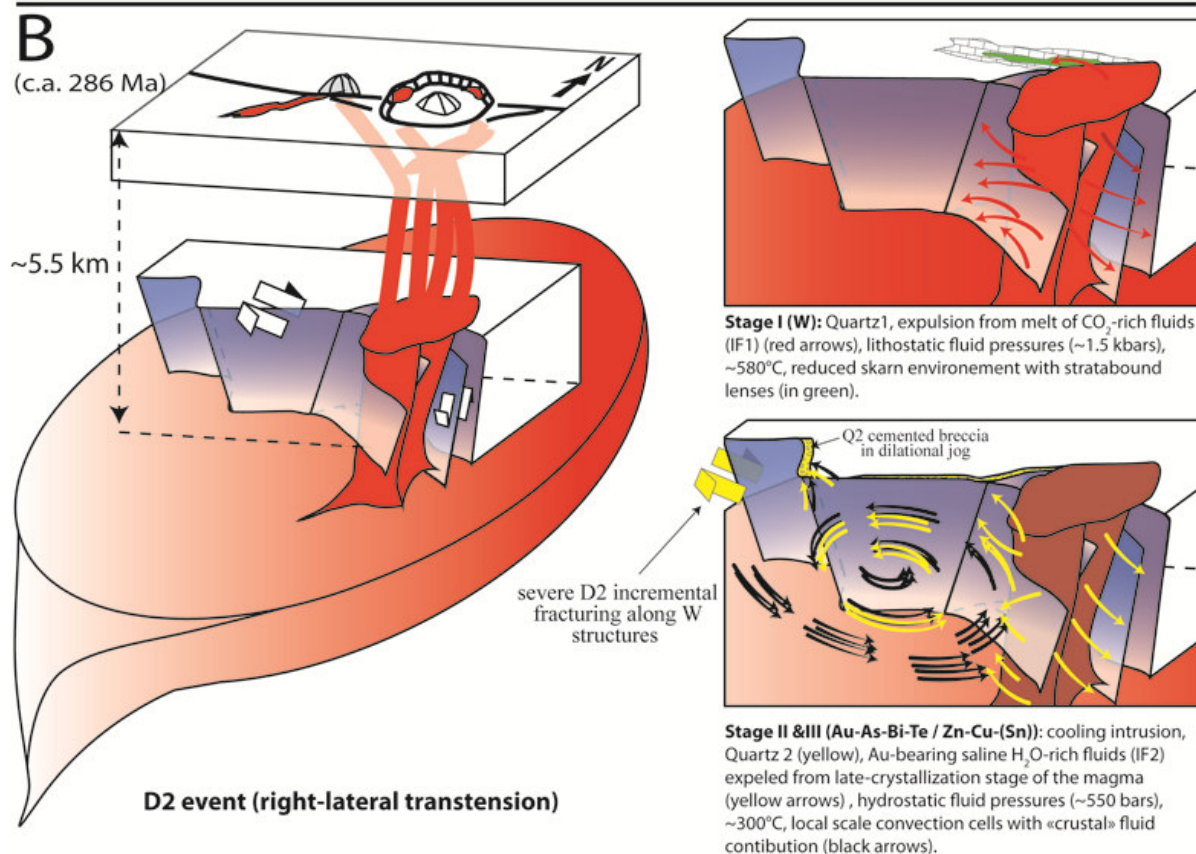
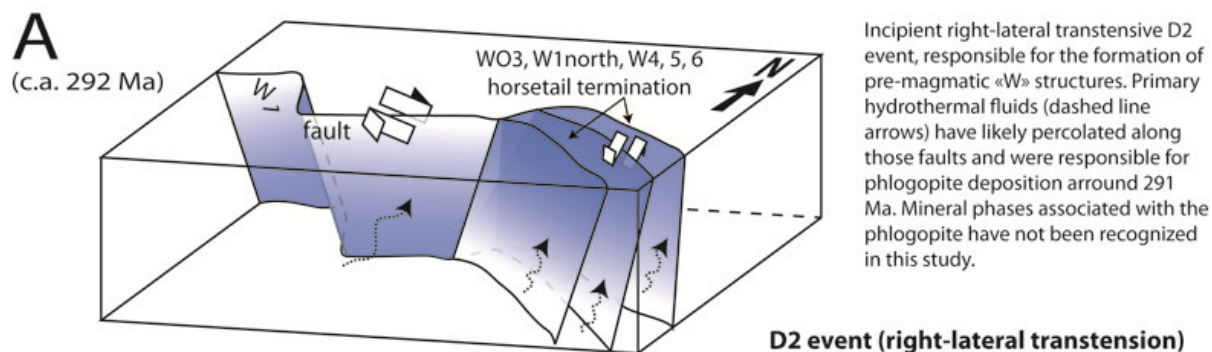












Highlights :

« Late-Hercynian Intrusion gold-related gold deposits : an integrated model on the Tighza polymetallic district, central Morocco

Study describes the first R-IRG deposit in North-Africa to our knowledge.

Detailed mineralogical, isotopic (Pb, Sr-Nd), chronological and fluid inclusions data are given.

Original structural context with pre-granitic corridors explained the shape of the deposits.

3D model is proposed, discarding a genetic relationship of Pb-Ag veins with gold deposits.

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